Methods in Dynamical Systems and Applications in Engineering

Lead Guest Editor: Fairouz Tchier Guest Editors: Ioannis Dassios and Lakhdar Ragoub



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Research Article

Time-Scale Integral Inequalities of Copson with Steklov Operator in High Dimension

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The paper derives some new time-scale (TS) dynamic inequalities for multiple integrals. The obtained inequalities are special cases of Copson integral using Steklov operator in (TS) version with high dimension. We prove the inequalities with several formulas for the operator and in different cases $m > \mu + 1$ and $m < \mu + 1$ for every $\mu \ge 1$, using time-scales (TSs) setting for integral properties, chain rules, Fubini's theorem, and Hölder's inequality.

1. Introduction

Equations and inequalities are the core of scientific study and have a great influence on a huge number of applications. A large number of physical phenomena and engineering studies have been analyzed and explained through equations and inequalities. For this reason, the study in this field developed rapidly and many types of inequalities and equations appeared. Dynamic inequalities on (TS) are some of the important inequalities that were extended by a lot of researchers and have interesting applications. Furthermore, dynamic inequalities are used to study the behaviour of dynamic equations.

Mathematical analysis has been the most important study in mathematics for the past three decades. Integral inequalities are one of the main studies and the core of mathematical analysis. In the 20th century, a significant part of science was numerical inequalities as the first composition to be released in 1934, through the published study by P'olya et al. [1]. This framework of inequalities played a vital role in the improvement processes and various applications of mathematics.

A large number of essential studies of integral inequalities appeared in the twentieth century, including pure and applied mathematics study. In 1920, Hardy produced the discrete Hardy inequality [2]. This inequality was also proved by himself in [3] (see also [4]), using the variations calculus to obtain the following inequality that is very valuable across both technological sciences and mathematics. If p > 1 and $h \ge 0$ in $(0, \infty)$ and

$$H(x) = \int_0^x h(t) \mathrm{d}t < \infty,$$

then

$$\int_{0}^{\infty} \left(\frac{H}{x}\right)^{p} \mathrm{d}x < \Lambda^{p} \int_{0}^{\infty} h^{p}(x) \mathrm{d}x, \qquad (1)$$

where $\Lambda = p(p-1)^{-1}$ is the best possible constant (BPC). Several important assessments and their implementation are done by inequality (1). Furthermore, the inequality is true in case $0 < a < b < \infty$,

$$\int_{a}^{b} \left(\frac{H}{x}\right)^{p} \mathrm{d}x < \Lambda^{p} \int_{a}^{b} h^{p}(x) \mathrm{d}x, \qquad (2)$$

where $0 < \int_0^b h^p(x) dx < \infty$. The classical inequality of Hardy declares that if p > 1 and h is nonnegative and measurable on (a, b), then (2) is valid except $h \equiv 0$ a.e. in (a, b), considering the (BPC).

Integral inequalities (3) and (4) are established in 1928 by Hardy [5].

Let *f* be a nonnegative measurable function on $(0, \infty)$:

Then,

$$\int_{0}^{\infty} x^{a-p} (Hh)^{p} (x) dx \le \left(\frac{p}{|p-a-1|}\right)^{p} \int_{0}^{\infty} x^{a} h^{p} (x) dx, \quad \text{for } p > 1.$$
(4)

Later, in 1976, Copson studied the integral inequalities ([6], Theorem 1, Theorem 3) as follows.

Let *h* and *v* be functions such that they are nonnegative measurable on $(0, \infty)$;

$$V(x) = \int_0^x v(t)dt,$$

(Ch)(x) $\leq \begin{cases} \int_0^x h(t)v(t)dt, & \text{for } c > 1, \\ \int_x^\infty h(t)v(t)dt, & \text{for } c < 1. \end{cases}$

Then,

$$\int_{0}^{\infty} V^{-c}(x)v(x)(Ch)^{p}(x)dx$$
$$\leq \left(\frac{p}{|c-1|}\right)^{p} \int_{0}^{\infty} V^{p-c}(x)v(x)dx, \quad \text{for } p \ge 1$$

Many papers included new extensions and generalizations for the inequalities above in more general settings. For instance, in 1979, some generalizations of Hardy-type inequality were proved by Chan [7]. Then, in 1992, Pachpatte [8] generalized the inequalities that were produced by Chan [7]. In 2005, P. Rehak used (TS) setting to extend Hardy's inequalities [9]. In 2015, Pachpatte's inequalities [8] were extended by Saker and O'Regan [10], with setting of (TSs). Later, some extensions of (TSs) Hardy inequalities were done for functions with high dimensions (see, for example, [11–14]).

In 2021, Albalawi and Khan generalized the main integral of Hardy and Copson inequalities, using the Steklov operator. The operator is defined in the following formulas with considering conditions in two cases (for more details, see [15]).

The aim of this paper is extending the study in [16] that was used for some new Hardy-type inequalities to obtain new special Copson inequalities with the Steklov operator (see [15]) in (TS) versions with high dimension. The results below are proved in two cases $m > \mu + 1$ and $m < \mu + 1$ by considering some general conditions that can be applied for any variable in the integral. To achieve this paper, we use (TSs) settings in integrals properties, chain rules, Hölder's inequality, and Fubini's theorem.

The paper takes the following structure: After introduction, the main concepts of (TSs) are presented in Section 2. Then, in Section 3, we generalized a class of Copson inequalities pertaining the Steklov operator with (TS) in high dimension. Lastly, conclusion of our results is presented.

2. Preliminaries and Lemmas on Time Scales

We state the main concepts of (TSs) that are used in this paper (for more details about (TS) calculus, see [17, 18]).

(TS) calculus in continuous case and discrete analysis was introduced by Hilger [19] in 1988. We denote to a subset (TS) of the real numbers \mathbb{R} by \mathbb{T} . Hence, the sets of numbers \mathbb{R} , \mathbb{Z} , and \mathbb{N} can be considered as (TSs).

Let $\sigma: \mathbb{T} \longrightarrow \mathbb{T}$ be a forward jump operator, such that $\sigma(t) \coloneqq \inf\{s \in \mathbb{T}: s > t\}$, while $\varsigma: \mathbb{T} \longrightarrow \mathbb{T}$ is the backward jump operator, given by $\varsigma(t) = \sup\{s \in \mathbb{T}: s < t\}$ for all $t \in \mathbb{T}$.

If $\sigma(t) > t$, then *t* is right-scattered, and if $\varsigma(t) < t$, *t* is leftscatted. In the case if points are right-scattered and leftscattered at the same time, then they will be isolated. The point *t* is right-dense if $t < \sup \mathbb{T}$ and $\sigma(t) = t$, while *t* is leftdense if $t > \inf \mathbb{T}$ and $\varsigma(t) = t$.

Let $g: \mathbb{T} \longrightarrow \mathbb{R}$ be a continuous function and if it satisfied the continuity at all right-dense points in \mathbb{T} and the limits of the left-sided exist (finite) at all left-dense points in \mathbb{T} , then g is known rd-continuous. We use $C_r(\mathbb{T}, \mathbb{R})$ to denote the space of all rd-continuous.

A function $g: \mathbb{T} \longrightarrow \mathbb{R}$ is Δ -differentiable at $t \in \mathbb{T}$, if there is a real number $\beta = g^{\Delta}(t)$ and for all $\varepsilon > 0$, there exists a neighbor. U of t satisfies

$$|q(\sigma(t)) - q(s) - \beta(\sigma(t) - s)| \le \varepsilon |\sigma(t) - s|, \text{ for all } s \in U.$$

The Δ -derivative of a function g in high order $n \in N$ is given by

$$g^{\Delta^{n}}(t) = \left(g^{\Delta^{n-1}}(t)\right)^{\Delta}.$$

If the Δ -derivative of $g^{\Delta^{n-1}}(t)$ exists, the following examples show that the delta derivative for every number set of (TSs).

If $\mathbb{T} = \mathbb{R}$, then

$$g^{\Delta}(t) = g' = \lim_{\Delta t \longrightarrow 0} \frac{g(t + \Delta t) - g(t)}{\Delta t}, \quad \text{for all } t \in \mathbb{T}.$$

If $\mathbb{T} = \mathbb{N}$, then

$$g^{\Delta}(t) = g(t+1) - g(t), \quad \text{for all } t \in \mathbb{T}.$$

Let $g: \mathbb{T} \longrightarrow \mathbb{R}$; if g is continuous at right-scattered t, then it is delta-derivative of the function g, given by

$$g^{\Delta}(t) = \frac{g(\sigma(t)) - g(t)}{\sigma(t) - t}$$

In the case of t is not right-scattered, then the derivative of g is given by

$$g^{\Delta}(t) = \lim_{s \longrightarrow t} \frac{g(\sigma(t)) - g(s)}{t - s}$$
$$= \lim_{s \longrightarrow \infty} \frac{g(t) - g(s)}{t - s}.$$

Here, the limit exists. Note that if $\mathbb{T} = \mathbb{R}$, we have

$$\sigma(t) = t,$$

$$g^{\Delta}(t) = g'(t).$$

If $\mathbb{T} = \mathbb{Z}$, we have

$$\sigma(t) = t + 1$$
$$g^{\Delta}(t) = \Delta g(t),$$
$$\int_{a}^{b} g(t)\Delta t = \sum_{t=a}^{b-1} g(t).$$

Lemma 1. Let $h; g: \mathbb{T} \longrightarrow \mathbb{R}$ be delta-differentiable. Then,

$$(hg)^{\Delta} = h^{\Delta}g + h^{\sigma}h^{\Delta}$$
$$= fg^{\Delta} + f^{\Delta}g^{\sigma},$$
$$\left(\frac{h}{g}\right)^{\Delta} = \frac{h^{\Delta}g - hg^{\Delta}}{gg^{\sigma}}.$$
(5)

The Cauchy integral of a delta-differential function of $g(g^{\Delta})$ is defined by

$$\int_{a}^{d} g^{\Delta}(t)\Delta t = g(d) - g(a), \quad \text{for } a, d \in \mathbb{T}$$

The time-scale integration by parts formula is given by

$$\int_{a}^{d} h(t)g^{\Delta}(t)\Delta t = h(t)g(t)]_{a}^{d} - \int_{a}^{d} h^{\Delta}(t)g^{\sigma()}\Delta t, \quad a, d \in \mathbb{T}.$$
(6)

The infinite integrals are defined by

$$\int_{a}^{\infty} g(t) \Delta t = \lim_{d \to \infty} \int_{a}^{d} g(t) \Delta t.$$

If $\mathbb{T} = \mathbb{R}$, we have

$$\int_{a}^{d} g(t)\Delta t = \int_{a}^{d} g(t) \mathrm{d}t.$$

If $\mathbb{T} = \mathbb{Z}$, we get

$$\int_{a}^{d} g(t) \Delta t = \sum_{t=a}^{d-1} g(t).$$

Lemma 2 (chain rule [16]). Assume a continuous function, w: $\mathbb{R} \longrightarrow \mathbb{R}$, a delta-differentiable, and w: $\mathbb{T} \longrightarrow \mathbb{R}$, on \mathbb{T}^c and a continuous differentiable $h: \mathbb{R} \longrightarrow \mathbb{R}$. Then, there exists $c \in [t, \sigma(t)]$ with

$$(h \circ w)^{\Delta}(t) = h'(w(c))w^{\Delta}(t).$$
⁽⁷⁾

Lemma 3 (dynamic Hölder inequality). Let $a, d \in \mathbb{T}$ and $h, w \in C_{rd}([a,d]_T), [0,\infty)$). If $p_1, p_2 > 1$ with $1/p_1 + 1/p_2 = 1$, then

$$\int_{a}^{d} h(t)w(t)\Delta t \leq \left(\int_{a}^{d} h^{p_{1}}(t)\Delta t\right)^{1/p_{1}} \left(\int_{a}^{d} w^{p_{2}}(t)\Delta t\right)^{1/p_{2}}.$$
 (8)

Theorem 4 (Fubini's theorem [20]). Let (Y, N, μ_{Δ}) and $(\Sigma, L, \gamma_{\Delta})$ be (TS) measure spaces with finite dimension. Consider $(Y \times \Sigma, N \times L, \mu_{\Delta} \times \gamma_{\Delta})$ as the measure space, where $N \times L$ is the σ -algebra product that is generated by $\{E \times F: E \in N, F \in L\}$ and

$$(\mu_{\Delta} \times \gamma_{\Delta})(E \times F) = \mu_{\Delta}(E)\gamma_{\Delta}(F).$$

Then, Fubini's theorem satisfied.

To be more accurate, if $\xi: \Upsilon \times \Sigma \longrightarrow \mathbb{R}$ is $(\mu_{\Lambda} \times \gamma_{\Lambda})$ —integrable,

$$\Psi(\gamma) = \int_{\Sigma} \xi(\gamma, \Pi) \Delta \Pi, \quad \text{exists for } \Pi \in \Upsilon,$$

and

$$\Psi(\Pi) = \int_{\Upsilon} \xi(\gamma, \Pi) \Delta \gamma, \quad \text{exists for } \gamma \in \Sigma$$

Then,

$$\int_{\Upsilon} \Delta \gamma \int_{\Sigma} \xi(\gamma, \Pi) \Delta \Pi = \int_{\Sigma} \Delta \Pi \int_{\Upsilon} \xi(\gamma, \Pi) \Delta \gamma$$

3. Main Results

A new (TS) version of Copson-type inequality with Steklov operator for multiple integrals is obtained in this section. We consider the nonnegative rd-continuous functions w_l , f_l , g_l , and v_l are Δ -integrable and defined integrals. Throughout this paper, we set $K(t_1, \ldots, t_k)$ as the Copson–Steklov-type operator considering the existence of the integral and also finite.

Theorem 5. Let \mathbb{T}_l be a (TS) and $a \in [0, \infty)_{\mathbb{T}_l}$, for $1 \le l \le k$ with $l, k \in \mathbb{N}$. In addition, let w_l , f_l , g_l , and v_l be nonnegative and rd-continuous functions on $[a, \infty)_{\mathbb{T}_l}$. Furthermore, assume there exist $\mu, \lambda \ge 1$ such that

$$\frac{w_l^{\Delta_l}(t_l)}{w_l^{\sigma}(t_l)} \leq \mu \frac{V_l^{\Delta_l}(t_l)}{V_l(t_l)},$$

and

$$\frac{g_l^{\Delta_l}(t_l)}{g_l^{\sigma}(t_l)} \leq \lambda \frac{F^{\Delta_l}(t_1,\ldots,t_k)}{F(t_1,\ldots,t_k)},$$

where $\Delta_l = \partial/\partial t_l$ for every l,

 $V_l(t_l) = \int_{a}^{t_l} v_l(s_l) \Delta s_l, \quad \text{with } V_l(\infty) = \infty, \text{ and } w_l(a) = 0,$ and

$$F(t_1,\ldots,t_k) \coloneqq \int_a^{t_1} \ldots \int_a^{t_k} \prod_{l=1}^k \frac{1}{g_l(s_l)} \frac{v_l(s_l)}{V_l(s_l)} f$$
$$\cdot (s_1,\ldots,s_k) \Delta s_1,\ldots,\Delta s_k.$$

Define the operator

$$K(t_1,...,t_k) = \prod_{l=1}^k g_l(t_l) F(t_1,...,t_k),$$
 (9)

Then

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1}, \dots, \Delta t_{k} \\
\leq \left(\frac{p(\lambda+1)}{m-(\mu+1)}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1}, \dots, \Delta t_{k-1}\right) \Delta t_{k},$$
(10)

.

where $p \ge 1$ and $m > \mu + 1$.

Proof. We write the left side of (10) as follows:

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} \Gamma_{k} \Delta t_{1}, \dots, \Delta t_{k-1}, \qquad (11)$$

where Γ_k is the *k*-term

$$\Gamma_k = \int_a^\infty w_k^\sigma(t_k) \frac{v_k(t_k)}{V_k^m(t_k)} (K^\sigma(t_1,\ldots,t_k))^p \Delta t_k.$$

Using formula (6) for integration by parts to compute Γ_k , we have

$$\Gamma_{k} = \int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1}, \dots t_{k}))^{p} \Delta t_{k}$$

$$= [z(t_{k})u(t_{k})]_{a}^{\infty} - \int_{a}^{\infty} u(t_{k})(z(t_{k}))^{\Delta_{k}} \Delta t_{k},$$
(12)

where $u^{\Delta_k}(t_k) = (v_k(t_k)/V_k^m(t_k))$ and then $u(t_k) = (-m + 1)V_k^{-m+1}(t_k)$ and $z^{\sigma}(t_k) = w_k^{\sigma}(t_k)(K^{\sigma}(t_1,\ldots,t_k))^p$, implying that $z(t_k) = w_k(t_k)(K(t_1,\ldots,t_k))^p$, and hence,

Assume $\lambda \ge 1$ such that

$$\frac{g_k^{\Delta_k}(t_k)}{g_k^{\sigma}(t_k)} \leq \lambda \frac{F^{\Delta_k}(t_1,\ldots,t_k)}{F(t_1,\ldots,t_k)},$$

where $F^{\Delta_k} = (\partial F / \partial t_k)$, and since $c_l \in [t_l, \sigma(t_l)]$, we have

$$z^{\Delta_k}(t_k) \leq w_k^{\Delta_k}(t_k) \left(K^{\sigma}(t_1,\ldots,t_k)\right)^p + p(\lambda+1)w_k(t_k)$$
$$\cdot \left(K^{\sigma}(t_1,\ldots,t_k)\right)^{p-1} g_k^{\sigma}(t_k) F^{\Delta_k}(t_1,\ldots,t_k).$$

Substituting the previous quantities in (12) and since $V_l(\infty) = \infty$ and $w_l(a) = 0$, then we have

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k} = \frac{1}{m-1} \int_{a}^{\infty} \frac{1}{V_{k}^{m-1}(t_{k})} w_{k}^{\Delta_{k}}(t_{k}) (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k} + \frac{p(\lambda+1)}{m-1} \int_{a}^{\infty} \frac{1}{V_{k}^{m-1}(t_{k})} w_{k}(t_{k}) (K^{\sigma}(t_{1},\ldots,t_{k}))^{p-1} g_{k}^{\sigma}(t_{k}) F^{\Delta_{k}}(t_{1},\ldots,t_{k}) \Delta t_{k}.$$

Assume $\mu \ge 1$ such that

ſ∞

$$\frac{w_l^{\Delta_l}(t_l)}{w_l^{\sigma}(t_l)} \leq \mu \frac{V_l^{\Delta_l}(t_l)}{V_l(t_l)}.$$

Then, we obtain

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k} = \frac{\mu}{m-1} \int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k} + \frac{p(\lambda+1)}{m-1} \int_{a}^{\infty} \frac{1}{V_{k}^{m-1}(t_{k})} w_{k}(t_{k}) (K^{\sigma}(t_{1},\ldots,t_{k}))^{p-1} g_{k}^{\sigma}(t_{k}) F^{\Delta_{k}}(t_{1},\ldots,t_{k})) \Delta t_{k}.$$

Since $F^{\Delta_k}(t_1,\ldots,t_k) = (f(t_1,\ldots,t_k)/g_k(t_k)) (v_k(t_k)/V_k(t_k))$, then we have

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p} \Delta t_{k} \leq \frac{p(\lambda+1)}{m-1-\mu} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} w_{k}(t_{k}) \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p-1} g_{k}^{\sigma}(t_{k}) \frac{f(t_{1},\ldots,t_{k})}{g_{k}(t_{k})} \Delta t_{k}.$$

Then, Hölder's inequality (8) with indices p and p/(p-1) can be applied:

$$\Gamma_{k} = \int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p} \Delta t_{k} \leq \left(\frac{p(\lambda+1)}{m-1-\mu}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{\left(w_{k}^{\sigma}(t_{k})\right)^{p-1}} f^{p}(t_{1},\ldots,t_{k}) \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \Delta t_{k}.$$

$$(13)$$

Substituting Γ_k in (11) and applying Fubini's Theorem 4, then we obtain the inequality

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \left(\frac{p(\lambda+1)}{m-(\mu+1)}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k-1}\right) \Delta t_{k}.$$

Corollary 6. If l = 1 in Theorem 5, inequality (10) becomes

$$\int_{a}^{\infty} w^{\sigma}(t) \frac{v(t)}{V^{m}(t)} \left(K^{\sigma}(t)\right)^{p} \Delta t \leq \left(\frac{p(\lambda+1)}{m-(\mu+1)}\right)^{p} \int_{a}^{\infty} \frac{v(t)}{V^{m}(t)} \frac{w^{p}(t)}{\left(w^{\sigma}(t)\right)^{p-1}} \left(\frac{g^{\sigma}(t)}{g(t)}\right)^{p} f^{p}(t) \Delta t.$$

$$\tag{14}$$

Corollary 7. If $\mathbb{T} = \mathbb{R}$ in Corollary 6, we obtain

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$$\int_{a}^{\infty} w(t) \frac{v(t)}{V^{m}(t)} K^{p}(t) \mathrm{d}t \leq \left(\frac{p(\lambda+1)}{m-(\mu+1)}\right)^{p} \int_{a}^{\infty} \frac{v(t)}{V^{m}(t)} w(t) f^{p}(t) \mathrm{d}t.$$
(15)

Remark 8. Assume $\mu = 0$ and $\beta > \lambda$ in Corollary 7; then, we have Corollary 3 in [15]

$$\int_{a}^{\infty} w(t) \frac{v(t)}{V^{m}(t)} K^{p}(t) dt$$

$$\leq \left(\frac{\beta p}{m-1}\right)^{p} \int_{a}^{\infty} w(t) \frac{v(t)}{V^{m}(t)} f^{p}(t) dt.$$
(16)

Theorem 9. Let \mathbb{T}_l be (TS) and $a \in [0, \infty)_{\mathbb{T}_l}$, for $1 \le l \le k$ with $l, k \in \mathbb{N}$. In addition, let w_l , f_l , g_l , and v_l be nonnegative and

rd-continuous functions on $[a, \infty)_{\mathbb{T}_l}$. Furthermore, assume there exist $\lambda, \mu \ge 1$ such that

$$\frac{w_l^{\Delta_l}(t_l)}{w_l^{\sigma}(t_l)} \ge \mu \frac{V_l^{\Delta_l}(t_l)}{V_l(t_l)},$$

and

$$\frac{g_l^{\Delta_l}(t_l)}{g_l^{\sigma}(t_l)} \ge \lambda \frac{F^{\Delta_l}(t_1,\ldots,t_k)}{F(t_1,\ldots,t_k)},$$

where $F^{\Delta_l} = (\partial F / \partial t_l)$; for every l,

$$V_l(t_l) = \int_a^{t_l} v_l(s_l) \Delta s_l, \quad \text{with } V_l(\infty) = \infty, \text{ and } w_l(a) = 0,$$

and

$$F(t_1,\ldots,t_k) \coloneqq \int_{t_1}^{\infty} \ldots \int_{t_k}^{\infty} \prod_{l=1}^k \frac{1}{g_l(s_l)} \frac{v_l(s_l)}{V_l(s_l)} f(s_1,\ldots,s_k) \Delta s_1 \ldots \Delta s_k$$

Define the operator

$$K(t_1,\ldots,t_k) = \prod_{l=1}^k g_l(t_l)F(t_1,\ldots,t_k)$$

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{\kappa} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \left(\frac{p(\lambda+1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k-1}\right) \Delta t_{k},$$
(17)

where $p \ge 1$ and $0 \le m < \mu + 1$.

Proof. We write the left side of (17) as follows:

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} \Gamma_{k} \Delta t_{1} \dots \Delta t_{k-1}.$$
 (18)

Use formula (6) to calculate the following k-term:

$$\Gamma_{k} = \int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k}$$
$$= \left[u(t_{k})z(t_{k})\right]_{a}^{\infty} + \int_{a}^{\infty} (u(t_{k}))(-z(t_{k}))^{\Delta_{k}} \Delta t_{k},$$
(19)

where $u^{\Delta_k}(t_k) = w_k^{\sigma}(t_k)(v_k(t_k)/(V_k(t_k))^m)$ and $z^{\sigma}(t_k) = (K^{\sigma}(t_1,\ldots,t_k))^p$.

Using (7) and the product rule (5), there exists $c_k \in [s_k, \sigma(s_k)]$ such that

$$\left(w_k(s_k) V_k^{1-m}(s_k) \right)^{\Delta_k} = w_k^{\Delta_k}(s_k) V_k^{1-m}(s_k) + w_k^{\sigma} (1-m) \left(V_k^{-m}(c_k) \right) V_k^{\Delta_k}(s_k).$$

Assume $\mu \ge 1$ such that

$$\frac{w_k^{\Delta_k}(t_k)}{w_k^{\sigma}(t_k)} \ge \mu \frac{V_k^{\Delta_k}(t_k)}{V_k(t_k)}.$$

Since $V_k^{\Delta_k}(s_k) = v_k(s_k) \ge 0$, $s_k \le c_k \le \sigma(s_k)$, and $0 \le m < 1$, then

Then,

$$w_{k}^{\sigma}V_{k}^{-m}(s_{k})v_{k}(s_{k}) \leq \frac{1}{1-m+\mu} \left(w_{k}(s_{k})V_{k}^{1-m}(s_{k})\right)^{\Delta_{k}}.$$
 (20)

By integration, we have

$$u(t_k) \leq \frac{1}{1-m+\mu} \int_a^\infty \left(w_k(s_k) V_k^{1-m}(s_k) \right)^{\Delta_k} \Delta s_k$$

Now, we calculate $(-K^p(t_1,\ldots,t_k))^{\Delta_k}$ and we obtain

$$(K^{p}(t_{1},\ldots,t_{k}))^{\Delta_{k}}$$

$$\cdot \left[g_{k}^{\Delta_{k}}(t_{k})F(t_{1},\ldots,t_{k})+g_{k}^{\sigma}(t_{k})F^{\Delta_{k}}(t_{1},\ldots,t_{k})\right].$$

Assume $\lambda \ge 1$ such that

$$\frac{g_k^{\Delta_k}(t_k)}{g_k^{\sigma}(t_k)} \ge \lambda \frac{F^{\Delta_k}(t_1,\ldots,t_k)}{F(t_1,\ldots,t_k)},$$

where

$$F(t_1,\ldots,t_k) \coloneqq g_k(t_k) \int_{t_k}^{\infty} \frac{1}{g_k(s_k)} \frac{\nu_k(s_k)}{V_k(s_k)} f(s_1,\ldots,s_k) \Delta s_k,$$
(21)

and since $V_k(\infty) = \infty$ and $c_k \ge s_k$, then we have

$$(K^{p}(t_{1},...,t_{k}))^{\Delta_{k}} \ge pK^{p-1}(t_{1},...,t_{k})(\lambda+1)g_{k}^{\sigma}(t_{k})F^{\Delta_{k}}(t_{1},...,t_{k})$$
$$\ge p(\lambda+1)K^{p-1}(t_{1},...,t_{k})\left(-\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\frac{v_{k}(t_{k})}{V_{k}(t_{k})}f(t_{1},...,t_{k})\right).$$

Then,

Hence, we have

$$(-K^{p}(t_{1},\ldots,t_{k}))^{\Delta_{k}}$$

$$\leq p(\lambda+1)K^{p-1}(t_{1},\ldots,t_{k})\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\frac{v_{k}(t_{k})}{V_{k}(t_{k})}f(t_{1},\ldots,t_{k}).$$

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p} \Delta t \leq \frac{p(\lambda+1)}{\mu+1-m} \int_{a}^{\infty} w_{k}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} f(t_{1},\ldots,t_{k}) \frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p-1} \Delta t_{k}.$$

Using Hölder's inequality, where $p_1 = p$ and $p_2 = (p/(p-1))$, we obtain

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p} \Delta t_{k} \leq \left(\frac{p(\lambda+1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{\left(w_{k}^{\sigma}(t_{k})\right)^{p-1}} f^{p}(t_{1},\ldots,t_{k}) \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \Delta t_{k}.$$

$$(22)$$

Substituting (22) in (17), we have

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \left(\frac{p(\lambda+1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}^{\sigma}(t_{k})}{g_{k}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k-1}\right) \Delta t_{k}.$$

Corollary 10. If l = 1 and $\mathbb{T} = \mathbb{R}$ in Theorem 9, we get

$$\int_{a}^{\infty} w(t) \frac{v(t)}{V^{m}(t)} K^{p}(t) dt$$

$$\leq \left(\frac{p(\lambda+1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v(t)}{V^{m}(t)} w(t) f^{p}(t) dt,$$
(23)

where

$$K(t) = g(t) \int_{t}^{\infty} \frac{1}{g(s)} \frac{v(s)}{V(s)} f(s) \mathrm{d}s.$$

Remark 11. Assume $\mu = 0$ and $\theta > \lambda$ in Corollary 10; we have Corollary 5 in [15].

Theorem 12. Let \mathbb{T}_l be (TS) and $a \in [0, \infty)_{\mathbb{T}_l}$, for $1 \le l \le k$ with $l, k \in \mathbb{N}$. In addition, let w_l , f_l , g_l , and v_l be nonnegative rd-continuous functions on $[a, \infty)_{\mathbb{T}_l}$. Furthermore, assume there exist $\lambda, \mu \ge 1$ such that

$$\frac{w_l^{\Delta_l}(t_l)}{w_l^{\sigma}(t_l)} \ge \mu \frac{V_l^{\Delta_l}(t_l)}{V_l(t_l)},$$

and

$$\frac{g_l^{\Delta_l}(t)}{g_l^{\sigma}(t)} \ge \lambda \frac{F^{\Delta_l}(t_1, \dots, t_k)}{F(t_1, \dots, t_k)}$$

where for every l,

$$V_l(t_l) = \int_a^{t_l} v_l(s_l) \Delta s_l$$
, with $V_l(\infty) = \infty$, and $w_l(a) = 0$,

and

$$F(t_1,\ldots,t_k) \coloneqq \int_{t_1}^{\infty} \ldots \int_{t_k}^{\infty} \prod_{l=1}^k g_l(s_l) \frac{v_l(s_l)}{V_l(s_l)} f(s_1,\ldots,s_k) \Delta s_1 \ldots \Delta s_k.$$

Define the operator

$$K(t_1,\ldots,t_k) \coloneqq \prod_{l=1}^k \frac{1}{g_l(t_l)} F(t_1,\ldots,t_k).$$

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \left(\frac{p(\lambda-1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1}, \dots, \Delta t_{k-1}\right) \Delta t_{k},$$

$$(24)$$

where $p \ge 1$ and $0 \le m < \mu + 1$.

Proof. We write the left side of (24) as follows:

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} \Gamma_{k} \Delta t_{1} \dots \Delta t_{k-1}.$$
(25)

Apply (6) to calculate the following k-term:

$$\Gamma_{k} = \int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},\ldots,t_{k}))^{p} \Delta t_{k}$$

$$= [u(t_{k})z(t_{k})]_{a}^{\infty} + \int_{a}^{\infty} (u(t_{k}))(-(z(t_{k}))^{\Delta_{k}} \Delta t_{k},$$
(26)

where $u^{\Delta_k}(t_k) = w_k^{\sigma}(t_k)(v_k(t_k)/V_k^m(t_k))$. Using the chain rule on (TS) (7) and product rule (5), there exist $c_k \in [s_k, \sigma(s_k)]$ such that

Then,

$$\left(w_k(s_k) V_k^{1-m}(s_k) \right)^{\Delta_k}$$

+ $(1-m) w^{\sigma}(s_k) V_k^{-m}(c_k) V_k^{\Delta_k}(s_k).$

Assume $\mu \ge 1$ such that

$$\frac{w_k^{\Delta_k}(t_k)}{w_k^{\sigma}(t_k)} \ge \mu \frac{V_k^{\Delta_k}(t_k)}{V_k(t_k)}.$$

Since $V_k^{\Delta_k}(s_k) = v_k(s_k) \ge 0$, $s_k \le c_k \le \sigma(s_k)$, and $0 \le m < \mu + 1$, then

$$(w_k(s_k)V_k^{1-m}(s_k))^{\Delta_k} \ge (1-m+\mu)w_k^{\sigma}V_k^{-m}(s_k)v_k(s_k),$$

implying

$$u(t_k)$$

$$\leq \frac{1}{1-m+\mu} \int_a^{t_k} \left(w_k(s_k) V_k^{1-m}(s_k) \right)^{\Delta_k} \Delta s_k$$

$$\leq \frac{1}{1-m+\mu} w_k(t_k) V_k^{1-m}(t_k).$$

We calculate $(-(K^p(t_1,\ldots,t_k))^{\Delta_k})$, and we obtain

$$(K^{p}(t_{1},\ldots,t_{k}))^{\Delta_{k}} = pK^{p-1}(t_{1},\ldots,c_{k}) \left[\frac{-g_{k}^{\Delta}}{g_{k}^{\sigma}(t_{k})g(t_{k})}F(t_{1},\ldots,t_{k}) + \frac{1}{g_{k}^{\sigma}(t_{k})}F^{\Delta_{k}}(t_{1},\ldots,t_{k}) \right].$$
Assume $\lambda \geq 1$ such that
$$F(t_{1},\ldots,t_{k}) \coloneqq \frac{1}{g_{k}(t_{k})} \int_{a}^{t_{k}} g_{k}(s_{k}) \frac{v_{k}(s_{k})}{V_{k}(s_{k})} f(s_{1},\ldots,s_{k}) \Delta s_{k},$$
(27)

$$\frac{g_k(t_k)}{g_k(t_k)} \ge \lambda \frac{F(t_1,\ldots,t_k)}{F(t_1,\ldots,t_k)},$$

and since $c_k \ge t_k$, then we have

where

$$(K^{p}(t_{1},...,t_{k}))^{\Delta_{k}} \ge pK^{p-1}(t_{1},...,t_{k})(1-\lambda)\frac{1}{g_{k}^{\sigma}(t_{k})}F^{\Delta_{k}}(t_{1},...,t_{k})$$

$$\ge p(\lambda-1)K^{p-1}(t_{1},...,t_{k})\frac{1}{g_{k}^{\sigma}(t_{k})}g_{k}(t_{k})\frac{\nu_{k}(t_{k})}{V_{k}(t_{k})}f(t_{1},...,t_{k}).$$

Since $V(\infty) = \infty$, then

$$(-K^{p}(t_{1},\ldots,t_{k}))^{\Delta_{k}} \leq -p(\lambda-1)K^{p-1}(t_{1},\ldots,t_{k})\frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})}\frac{v_{k}(t_{k})}{V_{k}(t_{k})}f(t_{1},\ldots,t_{k})$$

$$\leq p(\lambda-1)K^{\sigma(p-1)}(t_{1},\ldots,t_{k})\frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})}\frac{v_{k}(t_{k})}{V_{k}(t_{k})}f(t_{1},\ldots,t_{k}).$$

Hence, we have

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} (K^{\sigma}(t_{1},...,t_{k}))^{p} \Delta t_{k}$$

$$\leq [u(\infty)z(\infty) - u(a)z(a)] + \frac{p(\lambda - 1)}{\mu + 1 - m} \int_{a}^{\infty} w_{k}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} f(t_{1},...,t_{k}) \frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})} (K^{\sigma}(t_{1},...,t_{k}))^{p-1} \Delta t_{k}$$

$$\leq \frac{p(\lambda - 1)}{\mu + 1 - m} \int_{a}^{\infty} w_{k}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} f(t_{1},...,t_{k}) \frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})} (K^{\sigma}(t_{1},...,t_{k}))^{p-1} \Delta t_{k}.$$

Then, Hölder's inequality (8) can be applied with indices p and p/(p-1):

$$\int_{a}^{\infty} w_{k}^{\sigma}(t_{k}) \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \left(K^{\sigma}(t_{1},\ldots,t_{k})\right)^{p} \Delta t_{k} \leq \left(\frac{p(\lambda-1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{\left(w_{k}^{\sigma}\right)^{p-1}} f^{p}(t_{1},\ldots,t_{k}) \left(\frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})}\right)^{p} \Delta t_{k}.$$

$$(28)$$

Substituting (28) in (26), we have

 $\int_{a}^{\infty} w(t) \frac{v(t)}{V^{m}(t)} K^{p}(t) dt$

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} (K^{\sigma}(t_{1}, \dots, t_{k}))^{p} \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \left(\frac{p(\lambda-1)}{\mu+1-m}\right)^{p} \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{w_{k}^{p}(t_{k})}{(w_{k}^{\sigma}(t_{k}))^{p-1}} \left(\frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})}\right)^{p} \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f^{p}(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k-1}\right) \Delta t_{k}.$$

(29)

Corollary 13. If l = 1 and $\mathbb{T} = \mathbb{R}$ in Theorem 12, we obtain

 $\leq \left(\frac{p(\lambda-1)}{\mu+1-m}\right)^p \int_a^\infty \frac{v(t)}{V^m(t)} w(t) f^p(t) \mathrm{d}t,$

$$K(t) = \frac{1}{g(s)} \int_{a}^{t} g(t) \frac{v(s)}{V(s)} f(s) \mathrm{d}s.$$

Example 14. Choose $\mu = m$, $\lambda = 2$, and p = 1 in Theorem 12. Hence, we get

$$\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} K^{\sigma}(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k}$$

$$\leq \int_{a}^{\infty} \frac{v_{k}(t_{k})}{V_{k}^{m}(t_{k})} \frac{g_{k}(t_{k})}{g_{k}^{\sigma}(t_{k})} w_{k}(t_{k}) \left(\int_{a}^{\infty} \dots \int_{a}^{\infty} \prod_{l=1}^{k-1} w_{l}^{\sigma}(t_{l}) \frac{v_{l}(t_{l})}{V_{l}^{m}(t_{l})} f(t_{1}, \dots, t_{k}) \Delta t_{1} \dots \Delta t_{k-1} \right) \Delta t_{k}.$$
(30)

4. Conclusions

(TSs) calculus is used in this paper to prove special cases of (TS) Copson–Steklov-type inequalities with several variables. The obtained inequalities would be interesting to

apply in different fields of mathematics (functional spaces, partial differential equations, mathematical modeling). Furthermore, the inequalities can be discussed in calculus, discrete calculus, and quantum calculus. As a perspective, we propose to study these results for other kinds of operators and solve the singularity that appeared in Theorem 12 with case $m > \mu + 1$.

Data Availability

All data that support the findings of this study are included within the article.

Conflicts of Interest

The author declares no conflicts of interest.

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Research Article

Novel Analytical and Numerical Approximations to the Forced Damped Parametric Driven Pendulum Oscillator: Chebyshev Collocation Method

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In this work, some novel approximate analytical and numerical solutions to the forced damped driven nonlinear (FDDN) pendulum equation and some relation equations of motion on the pivot vertically for arbitrary angles are obtained. The analytical approximation is derived in terms of the Jacobi elliptic functions with arbitrary elliptic modulus. For the numerical approximations, the Chebyshev collocation numerical method is introduced for analyzing the equation of motion. Moreover, the analytical approximation and numerical approximation using the Chebyshev collocation numerical method and the MATH-EMATICA command Fit are compared with the Runge–Kutta (RK) numerical solution. Also, the maximum distance error to all obtained approximations is estimated with respect to the RK numerical solution. The obtained results help many authors to understand the mechanism of many phenomena related to the plasma physics, classical mechanics, quantum mechanics, optical fiber, and electronic circuits.

1. Introduction

The pendulum oscillator and some related equation have been used as a physical model to solve several natural problems related to bifurcations, oscillations, and chaos such as nonlinear plasma oscillations [1–9], Duffing oscillators [10–14], and Helmholtz oscillations [12], and many other applications can be found in [15–24]. There are few attempts for analyzing the equation of motion of the nonlinear damped pendulum taking the friction forces into account [25]. The approximate solution was obtained in form of the Jacobi elliptic functions. However, there are many others forces in addition to the friction force that affect the motion of the pendulum such as perturbed and periodic forces. These forces appear in different dynamic systems and cannot be neglected due to their great impact on the behavior of the oscillator. For instance, the unforced damped driven nonlinear pendulum equation/or the unforced damped parametric driven pendulum equation

$$\ddot{\varphi} + 2\beta\dot{\varphi} + \phi(t)\sin\varphi = 0, \qquad (1)$$

has been derived in detail in [26], where $\phi(t) = \omega_0^2 - \varepsilon \omega_2 \cos(\gamma t)$, $\omega_0^2 = g/l$, $\beta = \mu/2ml$, $\omega_1 = \gamma/ml$, $\omega_2 = \gamma^2/l$, $\varepsilon \ll 1$ is a small parameter, and $\varphi \equiv \varphi(t)$ denotes the angular displacement. In (1), ω_0 indicates the eigenfrequency of the system and β represents the damping coefficient. Here, the pendulum is modeled by a sphere of mass *m*, hanging at the end of a massless wire with length *l* and fixed to a supporting

point "O," swinging to and from in a vertical plane under the gravity acceleration "g.". For $(\beta, \omega_2, F) = (0, 0, 0)$, the unforced undamped nonlinear pendulum oscillation/the unforced undamped Duffing oscillator is recovered [5]. (1) has only been analyzed numerically via the midpoint scheme, and based on our comprehensive survey, we did not find any attempt to find a semi-analytical solution to this equation. Motivated by the potential applications of the nonlinear oscillators, the forced damped driven nonlinear (FDDN) pendulum equation or sometimes called the forced damped parametric driven pendulum equation will be studied:

$$\ddot{\varphi} + 2\beta\dot{\varphi} + \phi(t)\sin\varphi = F\cos(\Omega t).$$
(2)

Also, some analytical approximations to (2) and some related equations will be derived for the first time and will be compared with the Runge–Kutta (RK) numerical solution. Moreover, the Chebyshev collocation numerical method [27–29] is introduced for analyzing both (1) and (2). Furthermore, the MATHEMATICA command Fit is devoted for analyzing the equation of motion. We graphically make a comparison between the analytical and numerical approximations, and the maximum distance error in the whole time domain is estimated.

2. Analytical Approximations to the FDDN Pendulum Equation

Let us now write the evolution equation in the form of the initial value problem (i.v.p.):

$$\begin{cases} c\ddot{\varphi} + 2\beta\varphi + \phi(t)\sin\varphi = F\cos(\Omega t),\\ \varphi(0) = \varphi_0 \text{ and } \varphi'(0) = \dot{\varphi}_0.\\ 0 \le t \le T, \end{cases}$$
(3)

where $\varphi(t = 0) = \varphi_0$ indicates the oscillation amplitude.

Using Chebyshev polynomial approximation, we can approximate sin φ as

$$\sin \varphi \approx \varphi - \lambda \varphi^3 \text{ for } - M \le \varphi \le M, \tag{4}$$

where

$$\lambda \equiv \lambda_M = \frac{1}{6} + \frac{M}{5569} - \frac{M^2}{112} + \frac{M^3}{1883}.$$
 (5)

The error E_M of this approximation may be estimated via the following formula:

$$E_M = \frac{3}{298}M^3 - \frac{5}{378}M^2 + \frac{1}{203}M - \frac{1}{2837} \text{ for } \frac{2\pi}{180} \le M \le \frac{\pi}{2}.$$
(6)

For example, at the angle $M = 30^{\circ}$, the exact error equals $E_{ex} = 0.0000608542$ while the error according to formula (6) equals $E_M = 0.0000455313$ and the difference between them is given by $E = E_{ex} - E_M = 0.0000153229$. The respective approximation for $-30^{\circ} \le \varphi \le 30^{\circ}$ reads as

$$\sin \varphi \approx \varphi - 0.164389 \varphi^3 \approx \varphi - \frac{9}{55} \varphi^3.$$
 (7)

Also, for $M = 75^{\circ}$, we obtain $\lambda = 2/13$ which will be used as the default value in the present study. Consequently, i.v.p. (3) can be reduced to the following variable coefficient forced damped Duffing i.v.p.

$$\begin{cases} c\mathbb{Q} \equiv \ddot{\varphi} + 2\beta\dot{\varphi} + \phi(t)\left(\varphi - \frac{2}{13}\varphi^3\right) - F\cos\left(\Omega t\right) = 0,\\ \varphi(0) = \varphi_0 \text{ and } \varphi'(0) = \dot{\varphi}_0. \end{cases}$$
(8)

Suppose the solution of this problem is given by

$$\begin{cases} c\varphi = \theta + c_1 \cos(\Omega t) + c_2 \sin(\Omega t), \\ \theta(0) \equiv \theta_0 = \varphi_0 - c_1 \text{ and } \varphi'(0) \equiv \theta_1 = \dot{\varphi}_0 - c_2 \Omega. \end{cases}$$
(9)

The function $\theta \equiv \theta(t)$ is a solution to the following ode:

$$\ddot{\theta} + 2\beta \dot{\theta} + \phi(t) \left(\theta - \frac{2}{13}\theta^3\right) = 0.$$
(10)

Accordingly, we get

$$Q = -\frac{1}{26}\cos(\Omega t)A_1 - \frac{1}{26}\sin(\Omega t)A_2 -\frac{1}{26}(12\theta^2 A_3 + 6\theta A_4 + A_5)\phi(t),$$
(11)

where the coefficients $A_1 - A_5$ are given in Appendix 1.

Now, for small γ , we can define $\phi(t) = \omega_0^2 - \varepsilon \omega_2 \cos(\gamma t) \approx \omega_0^2 - \varepsilon \omega_2 = \kappa$ which leads to

$$Q \approx -\frac{1}{26}\cos(\Omega t)B_{1} - \frac{1}{26}\sin(\Omega t)B_{2} -\frac{\kappa}{26}(12\theta^{2}A_{3} + 6\theta A_{4} + A_{5}),$$
(12)

where the coefficients $A_1 - A_3$ have the same values given in Appendix 1 while the values of coefficients of B_1 and B_2 are given in Appendix 2.

The constants c_1 and c_2 could be determined from the following system:

$$\begin{cases} c3(c_1^3 + 3c_2^2c_1 - 26c_1)\kappa - 52c_2\beta\Omega + 26c_1\Omega^2 + 26F = 0, \\ (3c_2^3 + 3c_1^2c_2 - 26c_2)\kappa + 52c_1\beta\Omega + 26c_2\Omega^2 = 0. \end{cases}$$
(13)

Eliminating c_2 from system (13), we have the following cubic equation:

$$-26c_{1}\left(\frac{3F^{2}\kappa^{2}-3F^{2}\kappa\Omega^{2}-416\beta^{4}\Omega^{4}}{-104\beta^{2}\kappa^{2}\Omega^{2}+208\beta^{2}\kappa\Omega^{4}-104\beta^{2}\Omega^{6}}\right)+9F^{2}\kappa^{2}c_{1}^{3}$$

$$\left(-624F\beta^{2}\Omega^{2}c_{1}^{2}+78F^{3}-2704F\beta^{2}\Omega^{2}\right)\kappa+2704F\beta^{2}\Omega^{4}=0.$$
(14)

Also, by eliminating
$$c_1$$
 from system (13), we get
 $(10816\beta^4\Omega^4 + 2704\beta^2\kappa^2\Omega^2 - 5408\beta^2\kappa\Omega^4 + 2704\beta^2\Omega^6)c_2$
 $+9F^2\kappa^2c_2^3 + (312F\beta\kappa\Omega^3 - 312F\beta\kappa^2\Omega)c_2^2 - 5408F\beta^3\Omega^3 = 0.$
(15)

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We choose the least in magnitude pair of real roots (c_1, c_2) in (14) and (15). Accordingly, the final form of the analytical approximation to i.v.p. (3) is given by

$$\varphi_{\text{approx}}(t) = \theta(t) + c_1 \cos(t\Omega) + c_2 \sin(t\Omega), \quad (16)$$

with

$$\theta(t) = \frac{e^{-\beta t}}{1 + b_2 sn(f(t)\sqrt{\omega}|m)^2} \begin{pmatrix} b_1 dn(f(t)\sqrt{\omega}|m)sn(f(t)\sqrt{\omega}|m) \\ + \theta_0 cn(f(t)\sqrt{\omega}|m) \end{pmatrix},$$

$$f(t) = \frac{2\sqrt{-330\beta^2 - 329\kappa/\kappa\%}}{\sqrt{329}\gamma} E\left(\frac{t\gamma}{2} | \frac{658\varepsilon\omega_2}{330\beta^2 - 329\kappa}\right),$$
(17)

where

$$\omega = -\frac{p}{2m-1},$$

$$m = \frac{1}{2} \left(1 - \frac{p}{\sqrt{\left(p + q\varphi_0^2\right)^2 + 2\theta_1^2 q}} \right),$$

$$b_1 = \frac{\delta \theta_1 \sqrt{1-2m}}{\sqrt{p}},$$

$$b_2 = \frac{p + q\varphi_0^2 - \omega}{2\omega},$$

$$p = \kappa, q = -\frac{2}{13}\kappa.$$
(18)

3. Chebyshev Collocation Numerical Scheme for Analyzing FDDN Pendulum Equation

Now, the Chebyshev interpolation collocation method is introduced for analyzing i.v.p. (3) on the time interval [0, T]. To do that, we first solve numerically the ode. Let $\hat{\varphi}$ be the RK numerical solution to the i.v.p. described by (3). Then, we assume that the solution is given in terms of Chebyshev polynomials:

$$\varphi(t) = \sum_{k=0}^{n} c_k T_k \left(\frac{2t}{T} - 1\right),\tag{19}$$

where $T_k(t)$ stands for the Chebyshev polynomial of the first kind and *n* denotes the highest degree of the Chebyshev polynomials involved in the linear combination.

The collocation points t_k are defined as

$$t_{k} = \frac{1}{2}T\left[1 + \cos\left(\frac{4k+1}{2(n+1)}\pi\right)\right],$$
 (20)

for k = 1, 2, 3, ..., n.

The additional equations are required for determining the values of c_k . Thus, the values of the coefficients c_k are found from the following linear system:

$$\varphi(0) = \varphi_0,$$

$$\varphi'(0) = \dot{\varphi}_0,$$

$$\varphi''(0) = \hat{\varphi}''(0).$$

$$\varphi(T) = \hat{\varphi}(T),$$

$$\varphi'(T) = \hat{\varphi}'(T),$$

$$\varphi''(T) = \hat{\varphi}''(T).$$

$$\varphi(t_k) = \hat{\varphi}(t_k) \text{ for } k = 5, 6, \dots, n.$$

(21)

In general, increasing the value of n will not guarantee good approximations. Thus, we must choose some optimal value for n to our approximations. To this end, we define a range for possible n values, say

$$7 \le n_{\min} \le n \le n_{\max}.$$

We then find the optimal value for *n* within this range. Let $\varphi_n(t)$ be the solution using formula (19) and let $\varphi_{RK}(t)$ be the RK numerical solution to i.v.p. (3) on the interval $0 \le t \le T$. The following maximum distance error with respect to the RK numerical solution is defined:

$$E_{T,n} = \max_{0 \le t \le T} \left| \varphi_n(t) - \varphi_{\text{RK}}(t) \right|.$$
(23)

The optimal value for *n* on the range $n_{\min} \le n \le n_{\max}$ will then be that for which the error $E_{T,n}$ is as small as possible. *o* verifies the validity of this claim. Let us use the following data: $(\beta, \gamma, \omega_0, \omega_2, \varepsilon, \Omega, F, \varphi_0, \dot{\varphi}_0) = (0.2, 0.2, 1, 1, 0.2, 1, 0.1, 0, 0)$, as an example in i.v.p. (3). By solving this problem via both RK and Chebyshev collocation numerical methods in the interval $0 \le t \le 50$ and estimating the error $E_{T,n}$ based on relation (23) for $7 \le n \le 60$, we finally get the error $E_{T,n}$ associated with each number *n* as shown in Table 1. The results in Table 1 illustrate that the optimal value of *n* based on the mention data for $(\beta, \gamma, \omega_0, \omega_2, \varepsilon, \Omega, F, \varphi_0, \dot{\varphi}_0)$ equals n = 45 and the error value corresponding to n = 45 equals $E_{50,45} = 0.000964248$. Also, the optimal polynomial according to the mentioned data reads as

$$\begin{split} \varphi_{45}(t) &= 0.0511403t^2 - 0.0177057t^3 \\ &+ 0.0340205t^4 - 0.0823837t^5 + 0.109124t^6 - 0.0998239t^7 + 0.0677353t^8 \\ &- 0.0353909t^9 + 0.0146451t^{10} - 0.00490664t^{11} + 0.00135445t^{12} - 0.000312451t^{13} \\ &+ 0.0000609427t^{14} - 0.0000101493t^{15} + 1.45518 \times 10^{-6}t^{16} - 1.80894 \times 10^{-7}t^{17} \\ &+ 1.96141 \times 10^{-8}t^{18} - 1.86462 \times 10^{-9}t^{19} + 1.561 \times 10^{-10}t^{20} - 1.15515 \times 10^{-11}t^{21} \\ &+ 7.57989 \times 10^{-13}t^{22} - 4.42193 \times 10^{-14}t^{23} + 2.29822 \times 10^{-15}t^{24} - 1.06583 \times 10^{-16}t^{25} \\ &+ 4.41541 \times 10^{-18}t^{26} - 1.63491 \times 10^{-9}t^{27} + 5.41107 \times 10^{-21}t^{28} - 1.6 \times 10^{-22}t^{29} \\ &+ 4.22221 \times 10^{-24}t^{30} - 9.92634 \times 10^{-26}t^{31} + 2.07398 \times 10^{-27}t^{32} - 3.83854 \times 10^{-29}t^{33} \\ &+ 6.26651 \times 10^{-31}t^{34} - 8.97497 \times 10^{-33}t^{35} + 1.11996 \times 10^{-34}t^{36} - 1.2071 \times 10^{-36}t^{37} \\ &+ 1.1112 \times 10^{-38}t^{38} - 8.60994 \times 10^{-41}t^{39} + 5.50647 \times 10^{-43}t^{40} - 2.82885 \times 10^{-45}t^{41} \\ &+ 1.2272 \times 10^{-47}t^{42} - 3.22145 \times 10^{-50}t^{43} + 5.96106 \times 10^{-53}t^{44} - 5.33464 \times 10^{-56}t^{45}. \end{split}$$

Polynomial (24) allows us to estimate the cuts with the horizontal axis as well as the maxima and minima to the crest and the trough, respectively, as shown in Figure 1 and

Table 2. Using the MATHEMATICA command Fit gives the following solution $\varphi_{\text{Math}}(t) \equiv \varphi_{\text{Mathematica}}(t)$ for $(\beta, \gamma, \omega_0, \omega_2, \varepsilon, \Omega, F, \varphi_0, \varphi_0) = (0.2, 0.2, 1, 1, 0.2, 1, 0.1, 0, 0)$:

$$\varphi_{\text{Math}}(t) = e^{-t/5} \left(\frac{t^6}{365} - \frac{2t^5}{23} + \frac{125t^4}{94} - \frac{229t^3}{32} - \frac{21045t^2}{619} + \frac{33029t}{74} - \frac{63445}{76} \right) \sin(t) \\ + \frac{e^{-t/5}}{787321464} \left(\begin{array}{c} 2193096t^6 - 17115684t^5 - 426465793t^4 + 10891280252t^3 \\ -83720431296t^2 + 147163503646t + 393199198728 \end{array} \right) \cos(t) \\ + \frac{e^{-t/5}}{1544400} \left(\begin{array}{c} -3575t^7 + 93600t^6 - 1480050t^5 + 12725856t^4 \\ -38075400t^3 - 118006200t^2 + 1213821180t + 294883875 \end{array} \right) - \frac{11736}{17}, \end{array}$$

$$(25)$$

with error E = 0.000213725. Figure 2 demonstrates the comparison between the approximations of i.v.p. (3) using RK numerical solution with MATHEMATICA command Fit (here, polynomial (25)) and Chebyshev collocation numerical solution (24). It is noted that all used techniques give highly accurate approximations with low errors as compared to the RK numerical solutions.

The semi-analytical solution (16) to i.v.p. (3) could be recovered as follows.

Case (1). For $(\beta, \omega_2, F) = (0, 0, 0.1)$, the different approximations to the forced undamped Duffing oscillator with constant coefficients are introduced in Figure 3 with $(\gamma, \omega_0, \varepsilon, \Omega, \varphi_0, \dot{\varphi}_0) = (0.1, 1, 0.1, 2, 0, 0)$. The comparison between the RK method and the

analytical approximation (15) is presented in Figure 3(a). The approximate solutions using the MATHEMATICA command Fit and RK method are displayed in Figure 3(b). In Figure 3(c), both RK and Chebyshev collocation numerical approximations are presented. Also, the maximum error for the analytical approximation (15) and MATHEMATICA command Fit and Chebyshev collocation numerical solutions as compared to the RK numerical approximation is estimated based on the following relation:

$$E_{\infty}|_{\text{Type-solution}} = \max_{0 \le t \le 30} |\varphi_{\text{Type-solution}} - \varphi_{\text{RK}}|.$$
 (26)

Accordingly, the maximum error of the three approximations for the present case is estimated as

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N	ET:n
7	12:7318
8	8:18709
9	5:34902
10	3:73001
11	2:47806
12	5:99151
13	15:6472
14	19:1789
15	11:4247
16	12:9725
17	44:5793
Ν	ET;n
18	56:0277
19	22:7408
20	45:8666
21	86:0102
22	41:4478
23	54:48:00
24	81:8262
25	9:2908
26	50:9774
27	35:2609
28	7:88678
N	ET;n
29	24:0096
30	11:8633
31	11:9267
32	13:0699
33	16:4155
34	160:44:00
35	1021:51:00
36	455:30:00
37	293:27:00
38	62:37:00
39	594:34:00
N 40	E1;n
40	1011:59:00
41	1016:07:00
42	193:44:00
45	976:08:00
44	154:12:00
45	779.04.00
40	655:31:00
47	10/2:34:00
40	04.05.00
50	1060-25-00
N	FT·n
51	135:10:00
52	626.08.00
53	401.51.00
54	666:50:00
55	999:51:00
56	893:24:00
57	622:23:00
58	199:58:00
59	7:1974
60	30:729



FIGURE 1: The Chebyshev collocation and RK numerical approximations to i.v.p. (3) for $(\beta, \gamma, \omega_0, \omega_2, \varepsilon, \Omega, \varphi_0, \dot{\varphi}_0) = (0.2, 0.2, 1, 1, 0.2, 1, 0.1, 0, 0)$ and with n = 45 is plotted in the (φ, t) plane and the cuts with the horizontal axis as well as the maxima and minima to the crest and the trough is determined.

TABLE 2									
Zeros of the polynomial solution '45(t)				Zeros of the derivative '045(t)					
Κ	Tk	k	tk	k	tk	k	tk		
1	3:1993	11	34:9609	1	1:99978	11	26:6741		
2	6:39594	12	38:0328	2	4:93036	12	30:43:00		
3	9:46288	13	41:1493	3	7:98259	13	30:43:00		
4	12:4583	14	43:9479	4	10:9844	14	33:3364		
5	15:4562	15	46:912	5	13:9641	15	36:4941		
6	18:5134	16	49:9492	6	13:9645	16	36:4941		
7	21:6812			7	16:98	17	39:5245		
8	24:8139			8	20:0903	18	42:4811		
9	28:3547			9	23:3291	19	45:4226		
10	31:7165			10	26:674	20	48:4281		



FIGURE 2: The comparison between the approximations of i.v.p. (3) using RK numerical solution with MATHEMATICA command Fit (here, polynomial (25)) and Chebyshev collocation numerical solution (24) for $(\beta, \gamma, \omega_0, \omega_2, \varepsilon, \Omega, \varphi_0, \dot{\varphi}_0) = (0.2, 0.2, 1, 1, 0.2, 1, 0.1, 0, 0)$ and n = 45.



FIGURE 3: The comparison between the semi-analytical solution (analytical approximation) (16) and the numerical approximations using the Chebyshev collocation method and RK numerical method as well as the MATHEMATICA command Fit to i.v.p. (3) for case (1): $(\beta, \omega_2, F) = (0, 0, 0.1)$.

$$E|_{semi-analy} = 0.000214826,$$

$$E|_{Mathematica} = 0.00856153,$$
 (27)

$$E_{n=47}|_{Chebyshev} = 0.0000389093.$$

Case (2). For $(\beta, \omega_2, F) = (0, 1, 0.1)$, the comparison between the analytical approximation (16) and the numerical approximations using the RK, MATHE-MATICA command Fit, and Chebyshev collocation numerical methods to the forced undamped Duffing equation with variable coefficients is considered as shown in Figure 4 with $(\gamma, \omega_0, \varepsilon, \Omega, \varphi_0, \dot{\varphi}_0) =$ (0.1, 1, 0.1, 2, 0, 0). The maximum error of the three approximations to the present case is calculated as

$$E|_{semi-analy} = 0.00344896,$$

 $E|_{Mathematica} = 0.0100608,$ (28)
 $E_{n=47}|_{Chebyshev} = 0.0000237867.$

Case (3). For $(\beta, \omega_2, F) = (0.1, 1, 0)$, the unforced damped Duffing equation with variable coefficients is recovered and its semi-analytical solution (16) is compared with the numerical approximations using RK, MATHEMATICA command Fit, and Chebyshev collocation numerical methods as demonstrated in Figure 5 with $(\gamma, \omega_0, \varepsilon, \Omega, \varphi_0, \dot{\varphi}_0) = (0.1, 1, 0.1, 2, 0, 0.1)$. In addition, the maximum error to the three approximations as compared to RK numerical approximation is estimated as

$$E|_{semi-analy} = 0.000447172,$$

$$E|_{Mathematica} = 0.00133459,$$

$$E_{n=47}|_{Chebyshev} = 2.05447 \times 10^{-6}.$$

(29)

The MATHEMATICA code for the RK and Chebyshev collocation numerical approximations with the maximum error is given in Appendix 3.



FIGURE 4: The comparison between the analytical approximation (16) and the numerical approximations using the Chebyshev collocation method and RK numerical method as well as the MATHEMATICA command Fit to i.v.p. (3) for case (2): $(\beta, \omega_2, F) = (0, 1, 0.1)$.







FIGURE 5: The comparison between the analytical approximation (16) and the numerical approximations using the Chebyshev collocation method and RK numerical method as well as the MATHEMATICA command Fit to i.v.p. (3) for case (3):(β , ω_2 , F) = (0.1, 1, 0).



FIGURE 6: The comparison between the analytical approximation (16) and the numerical approximations using the Chebyshev collocation method and RK numerical method as well as the MATHEMATICA command Fit to i.v.p. (3) for case (4): $(\beta, \omega_2, F) = (0.1, 1, 0.1)$.



FIGURE 7: The analytical approximation (16) to i.v.p. (3) plotted in the (φ , *t*)plane for different values to the coefficient of the damping term β .

Case (4). For $(\beta, \omega_2, F) = (0.1, 1, 0.1)$, the general analytical approximation (16) to the forced damped parametric driven pendulum i.v.p. (3) is compared with the RK, MATHEMATICA command Fit, and Chebyshev collocation numerical solutions as elucidated in Figure 6. The influence of β on the amplitude of the semi-analytical solution (16) is investigated as shown in Figure 7. It is noted that the amplitude of the analytical approximation (15) decreases with the increase of β . Also, the maximum error according to relation (23) to the analytical approximation (16) and MATHEMATICA command Fit, and Chebyshev collocation numerical solutions as compared to the RK numerical solution is estimated as follows:

$$E|_{semi-analy} = 0.00326255,$$

 $E|_{Mathematica} = 0.0128206,$ (30)
 $E_{n=47}|_{Chebyshev} = 0.000155082.$

.

In all mentioned cases, it is observed that analytical approximation (16) to i.v.p. (3) and its related equations (here we mean the four mentioned cases) give highly accurate results as compared to the numerical approximations. It is observed that analytical approximation (16) is better than the MATHEMATICA approximation but less than the Chebyshev collocation numerical solution for all mentioned cases. However, in the fourth case, i.e., the forced damped parametric driven pendulum i.v.p. (3), the MATHEMA-TICA approximation is better than the analytical approximation (16). In general, all obtained approximations are characterized by their high accuracy. However, semi-analytical solution (16) is more stable than the Chebyshev collocation numerical solution against all relevant physical variables.

4. Conclusions

In this work, some effective and accurate analytical and numerical approximations to the forced damped parametric driven pendulum equation have been derived and investigated. The mentioned equation of motion has been reduced to the forced damped Duffing equation with variable coefficients in order to find its analytical solution. In terms of the Jacobi elliptic functions, the analytical approximation has been derived. For the numerical approximations, the Chebyshev collocation method has been used for analyzing the equation of motion and some related equations. It was noted that the analytical approximation could recover some special cases to the nonlinear pendulum oscillators. For instance, for undamping case, i.e., for $\beta = 0$, the solution to the forced undamped Duffing equation with variable coefficients has been recovered and examined. Also, for $(\beta, \omega_2) = (0, 0)$, the solution to the forced undamped Duffing equation with constant coefficients has been recovered and discussed. The obtained approximations were compared with the RK numerical approximation and the MATHEMATICA command Fit approximation. Also, the maximum distance error has been estimated for all approximations as compared to the RK numerical approximation. It was found that the analytical approximation gives good results with high accuracy as compared to the numerical approximations. Furthermore, it was observed that the analytical approximation is better than the MATHEMATICA approximation but less than the Chebyshev collocation numerical solution for all mentioned cases except the case of the forced damped parametric driven pendulum i.v.p. (3), the MATHEMATICA approximation is slowly better than the analytical approximation (16). The methods used in this study could be extended to solve many nonlinear equations that control the different cases of pendulum oscillations [30-33]. In addition, the obtained results/ solutions are useful for investigating several physical problems related to the oscillations in plasma physics, fluid mechanics, field theory, engineering science, solid state physics, and quantum mechanics.

Appendices

Appendix A

The coefficients $A_1 - A_5$ of equation (11):

$$\begin{split} A_{1} &= \begin{bmatrix} \phi(t) (3c_{1}^{3} + 3c_{2}^{2}c_{1} - 26c_{1}) \\ -52c_{2}\beta\Omega + 26c_{1}\Omega^{2} + 26F \end{bmatrix}, \\ A_{2} &= \begin{bmatrix} \phi(t) (3c_{2}^{3} + 3c_{1}^{2}c_{2} - 26c_{2}) \\ + (52c_{1}\beta\Omega + 26c_{2}\Omega^{2}) \end{bmatrix}, \\ A_{3} &= (c_{2}\sin(\Omega t) + c_{1}\cos(\Omega t)), \\ A_{4} &= \begin{bmatrix} (c_{1}^{2} - c_{2}^{2})\cos(2\Omega t) \\ + c_{1}^{2} + c_{2}^{2} + 2c_{2}c_{1}\sin(2\Omega t) \end{bmatrix}, \\ A_{5} &= [c_{2} (3c_{1}^{2} - c_{2}^{2})\sin(3\Omega t) + c_{1} (c_{1}^{2} - 3c_{2}^{2})\cos(3\Omega t)]. \end{split}$$
(A.1)

Appendix **B**

The coefficients B_1 and B_2 of equation (12):

$$B_{1} = \begin{bmatrix} \kappa (3c_{1}^{3} + 3c_{2}^{2}c_{1} - 26c_{1}) \\ -52c_{2}\beta\Omega + 26c_{1}\Omega^{2} + 26F \end{bmatrix},$$

$$B_{2} = \begin{bmatrix} \kappa (3c_{2}^{3} + 3c_{1}^{2}c_{2} - 26c_{2}) \\ + (52c_{1}\beta\Omega + 26c_{2}\Omega^{2}) \end{bmatrix}.$$
(B.1)

Appendix C

MATHEMATICA Code for Chebyshev Collocation Numerical Method to Figure 5(c). Note that this is general code which can be used and applied for analyzing many oscillators related to the present evolution equation.

Clear[a, b, m, h, x, \ [CurlyPhi], n, \ [Chi]]; $\{a = 0, b = 30, m = 45\}$;

x1,

y, \{t, 0, 100\}][[1, 1, 2]];

 $Plot[Evaluate[\{ rk[t] \}], \{ t, a, b - 10 \}, PlotRange - >rbin All.$

PlotStyle - > $rbin \{\{Black, Thin\}\}}$

h = (b - a)/m;

x[t_]:= Sum[

Subscript[c, k] ChebyshevT[k, (a + b - 2t)/(a - b)], \{k, 0, m\}];

 $R[t_] := x^{"}[t] + 2 \setminus [Beta] x^{'}[t] + \langle [Phi][t]^* Sin[x[t]] -$ F Cos[\setminus [CapitalOmega] *t*]; |Xi|[i]| = 1/2 (a + b + (-a + (b) Cos[(|Pi] + 4 i |Pi])/(2 + 2 m)])solc = Flatten[Solve[sys0]]; $x[t_] := Sum[$ Subscript[c, k] ChebyshevT[k, (a + b - 2t)/(a - b)]//. solc, \{k. 0, m]; $\[Chi][m][t]:=$ Sum[Subscript[c, k] ChebyshevT[k, (a + b - 2t)/(a - b)]//solc, $\{k, 0, m\}$; $er[m_] := Max[Table[Abs[rk[t] - \ [Chi][m][t]], \t, a, b,$ 0.1]; $\operatorname{error}[m_] := \operatorname{Module}[\{ \ [Xi], \ [Chi], R, sys, solc, err \},$ $\{ \ [Xi][j_]:=$ $1/2 (a + b + (-a + (b) \cos[([Pi] + 4j [Pi])/(2 + 2m)]);)$ [Chi][tt_]:= Sum[Subscript[c, k] ChebyshevT[k, (a + b - 2 tt)/(a - b)], $\{k, 0, \}$ m]; $R[t_]: = \ [Chi]"[tt] +$ $2 \setminus [Beta] \setminus [Chi]'[tt] + \setminus [Phi][tt] * Sin[\setminus [Chi][tt]] -$ F Cos[\ [CapitalOmega] tt]; sys = Flatten[Join] Table[rk[$[Xi][j] = [Chi][[Xi][j]], \{j, 5,$ m\}], \{\{ \ [Chi][a] - x0 = = 0, \ [Chi]'[a] - x1 = = 0, \ $[Chi][b] - rk[b] = = 0, \ [Chi]'[b] - rk'[b] = =$ 0, $[Chi]^{n}[b] - rk^{n}[b] = = 0]];$ solc = Flatten[NSolve[sys]]; \ [Chi][tt_]: = Sum[Subscript[c, k] ChebyshevT[k, (a + b - 2t)/(a - b)], \{k, 0, m]//. solc; $err = Maximize[\{Abs[rk[t] - [Chi][t]], 0 = t = b\}, t][[1]];$ \}; $\{err, \in [Chi][t]\}$]; $n_{\min} = 7; n_{\max} = 60;$ opt = Sort[Table[\{error[jj][[1]], jj\}, \{jj, n_{\min} , n_{\max} \}]] ||1||; n = opt[[2]];OPtimalnValue = = nerrx = error[n]; Error = errx[[1]]poly = errx[[2]]; $Cheb = Plot[Evaluate[\{ rk[t], poly \}], \{t, a, b\}, Plo$ tRange - > All. PlotStyle - > \{\{Dashing[0.05], Thick, Black\}, \{Dotted, Thick, Blue \} \}, PlotLegends - > Placed[\{"RK4", "Chebyshev"\}, Frame - > True].

Data Availability

All data generated or analyzed during this study are included within the article (more details can be requested from the corresponding author).

Disclosure

This paper was only published as a preprint entitled Approximate Solutions to the Forced Damped Parametric Driven Pendulum Oscillator: Chebyshev Collocation Numerical Solution [34].

Conflicts of Interest

The authors declare that they have no conflicts of interest.

Authors' Contributions

All authors contributed equally and approved the final version of the manuscript.

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Research Article

Fourth-Order Hankel Determinants and Toeplitz Determinants for Convex Functions Connected with Sine Functions

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This article deals with the upper bound of fourth-order Hankel and Toeplitz determinants for the convex functions which are defined by using the sine function. The main tools in this study are the coefficient inequalities for the class *P* of functions with positive real parts. Also, the investigation of the upper bound of the fourth-order Hankel determinant for 3-fold symmetric convex functions associated with the sine function is included.

1. Introduction

Let the family of all functions f be denoted by A which are analytic in an open unit disc $D = \{z \in \mathbb{C}: |z| < 1\}$ with Taylor series expansion:

$$f(z) = z + \sum_{n=2}^{\infty} a_n z^n, \, (z \in D),$$
(1)

and *S* represent a family of functions $f \in A$ which are univalent in *D*. Let *S*^{*}, *C*, and K_g denote the families of starlike, convex, and close-to-convex functions, respectively, and they are defined as

$$S^* = \left\{ f \in S: \, \Re\left(\frac{zf'(z)}{f(z)}\right) > 0, \, (z \in D) \right\},$$
$$C = \left\{ f \in S: \, \Re\left(1 + \frac{zf''(z)}{f'(z)}\right) > 0, \, (z \in D) \right\}, \tag{2}$$

$$K_g = \left\{ f \in S: \, \Re\left(\frac{zf'(z)}{g(z)}\right) > 0, \quad \text{for } g \in S^*, \, (z \in D) \right\}.$$

Let *P* denote the family of all analytic functions p of the form

$$p(z) = 1 + \sum_{n=1}^{\infty} c_n z^n, \ z \in \mathbb{D},$$
 (3)

with the positive real parts in *D*. As the n^{th} coefficient for the functions belonging to the family is bounded by *n*, this bound helps in the study of geometric properties of functions $f \in S$. Specifically, the second coefficient a_2 helps in finding the distortion and growth properties of a normalized univalent function. Likewise, the problems involving power series with integral coefficients and investigating the singularities are successfully handled by using Hankel determinants. Pommerenke [1, 2] introduced the idea of Hankel determinants, and he defined those for univalent functions $f \in S$ of form (7) as follows:

$$H_{q,n}(f) = \begin{vmatrix} a_n & a_{n+1} & \dots & a_{n+q-1} \\ a_{n+1} & a_{n+2} & \dots & a_{n+q} \\ \vdots & \vdots & \dots & \vdots \\ a_{n+q-1} & a_{n+q} & \dots & a_{n+2q-2} \end{vmatrix}.$$
 (4)

In the theory of analytic functions, finding the upper bound of $|H_{q,n}(f)|$ is one of the most studied problems. Several researchers found the above-mentioned bound for different subfamilies of univalent functions for fixed values of q and n. A few remarkable contributions in this regard are included here for reference. For the subfamilies S^* , C, and $K_z = R$ (the class of functions with bounded turnings) of the set S, the sharp bounds of $|H_{(2,2)}(f)|$ were investigated by Janteng et al. [3, 4]. They proved the bounds as follows:

$$|H_{(2,2)}(f)| \leq \begin{cases} 1 \text{ for } f \in S^*, \\ \frac{1}{8} \text{ for } f \in C, \\ \frac{4}{9} \text{ for } f \in R. \end{cases}$$
(5)

The accurate estimate of $|H_{(2,2)}(f)|$ was obtained by Krishna et al. [5] for the family of Bazilevic functions. For subfamilies of S, more studies regarding $H_{(2,2)}(f)$ can be seen in [6-12]. According to Thomas' conjecture [13], if $f \in S$, then $|H_{(q,2)}(f)| \le 1$, but it was shown by Li and Srivastava in [14] that this conjecture is not true for $n \ge 4$. Also, Raducanu and Zaprawa [15] showed that it is false for n = 2. Rather, they showed that $\max\{|H_{(2,2)}(f)|:$ $f \in S$ \geq 1.175. As compared to $|H_{(2,2)}(f)|$, estimation of $|H_{(3,1)}(f)|$ is much more difficult. Babalola [16] published the first paper on $H_{(3,1)}(f)$ in 2010 in which he obtained the upper bound of $|H_{(3,1)}(f)|$ for subfamilies of S^* , C, and R. After that, for different subfamilies of analytic and univalent functions, few other authors [17-25] also published their work regarding $|H_{(3,1)}(f)|$. Zaprawa [26] improved the results of Babalola [16] recently in 2017, by showing

$$|H_{(3,1)}(f)| \leq \begin{cases} 1 \text{ for } f \in S^*, \\ \frac{49}{540} \text{ for } f \in C, \\ \frac{41}{60} \text{ for } f \in R. \end{cases}$$
(6)

He claimed that these bounds are not sharp. Furthermore, he considered the subfamilies of S^* , C, and R for sharpness, having functions with *m*-fold symmetry, and obtained the sharp bounds. Arif et al. [27–30] made a remarkable contribution in studying the fourth- and fifth-order Hankel determinants $H_{(4,1)}(f)$ and $H_{(5,1)}(f)$ for certain subfamilies of univalent functions. Mashwani et al. [31] have studied the fourth-order Hankel determinant for starlike functions related to sigmoid functions, whereas Kaur et al. [32] studied the same problem for a subclass of bounded turning functions. Wang et al. [33] studied the problem for bounded turning functions related to the lemniscate of Bernoulli. Recently, Zhang and Tang [34] have studied the fourth-order Hankel determinant for the class of starlike functions related to sine functions. Motivated by the

above-mentioned work, we intend to add some contributions to the fourth-order Hankel determinant for the class of convex functions associated with sine functions. Recently, the following class C_s of convex functions was introduced, which is associated with the sine function:

$$C_{s} = \left\{ f \in A: \frac{(zf'(z))'}{f'(z)} < 1 + \sin z \ (z \in D) \right\},$$
(7)

where \prec is a subordination symbol and it also implies that the region defined by (zf'(D))'/f'(D) lies in the eightshaped region in the right-half plane. For different subfamilies of univalent functions, growth of $H_{q,n}(f)$ has been studied for fixed values of q and n. Particularly, we have

$$H_{4,1}(f) = \begin{vmatrix} a_1 & a_2 & a_3 & a_4 \\ a_2 & a_3 & a_4 & a_5 \\ a_3 & a_4 & a_5 & a_6 \\ a_4 & a_5 & a_6 & a_7 \end{vmatrix} (n = 1, q = 4).$$
(8)

Also, Thomas and Halim defined the symmetric Toeplitz determinant $T_q(n)$ as follows:

$$T_{q,n}(f) = \begin{vmatrix} a_n & a_{n+1} & \dots & a_{n+q-1} \\ a_{n+1} & a_n & \dots & a_{n+q} \\ \vdots & \vdots & \dots & \vdots \\ a_{n+q-1} & a_{n+q} & \dots & a_n \end{vmatrix} (n \ge 1, q \ge 1).$$
(9)

The Toeplitz determinants are closely related to Hankel determinants. As Hankel matrices consist of constant entries along the reverse diagonal, the Toeplitz matrices consist of constant entries along the diagonal.

As a special case, when n = 1 and q = 4, we have

$$T_{4,2}(f) = \begin{vmatrix} a_1 & a_2 & a_3 & a_4 \\ a_2 & a_1 & a_4 & a_3 \\ a_3 & a_4 & a_1 & a_2 \\ a_4 & a_3 & a_2 & a_1 \end{vmatrix}.$$
 (10)

In this paper, we intend to find the upper bound of $|H_{4,1}(f)|$ and $|T_{4,2}(f)|$ for the class of functions defined by (6). The following sharp results would be useful for investigating our main results.

Lemma 1. If $p \in P$ and p is of form (2), then for each $n, k, m, l \in \mathbb{N} = \{1, 2, ...\}$, the following sharp inequalities hold:

(

$$\begin{aligned} \left|c_{n}\right| &\leq 2,\\ c_{n+k} - \mu c_{n} c_{k}\right| &\leq \begin{cases} 2, \text{ for } 0 \leq \mu \leq 1\\ 2|2\mu - 1|, \text{ elsewhere} \end{cases}$$
(11)

$$c_{n+2k} - c_n c_k^2 \le 6,$$
 (12)

$$|c_n c_k - c_m c_l| \le 4$$
, for $n + k = m + l$. (13)

Inequalities (10)-(12) are proved in [26, 35, 36], respectively. Inequality (13) is obvious.

Libera and Złotkiewicz proved the following result [37].

Lemma 2. Let $p \in P$ be of form (2). Then, the modulus of the expressions

$$A_{3} = c_{1}^{3} - 2c_{1}c_{2} + c_{3},$$

$$A_{4} = c_{1}^{4} + c_{2}^{2} + 2c_{1}c_{3} - 3c_{1}^{2}c_{2} - c_{4},$$

$$A_{5} = c_{1}^{5} + 3c_{1}c_{2}^{2} + 3c_{1}^{2}c_{3} - 4c_{1}^{3}c_{2} - 2c_{1}c_{4} - 2c_{2}c_{3} + c_{5},$$

$$(14)$$

$$A_{6} = c_{1}^{6} + 6c_{1}^{2}c_{2}^{2} + 4c_{1}^{3}c_{3} + 2c_{1}c_{5} + 2c_{2}c_{4} + c_{3}^{2}$$

$$- c_{2}^{3} - 5c_{1}^{4}c_{2} - 3c_{1}^{2}c_{4} - 6c_{1}c_{2}c_{3} - c_{6}$$

are all bounded by 2.

2. Main Results

2.1. Bounds of $|H_{4,1}(f)|$ and $|T_{4,2}(f)|$ for the Set C_s Connected with the Sine Function. Following (7), we can write $H_{4,1}(f)$, where $f \in S$ and $a_1 = 1$, as

$$H_{4,1}(f) = a_7 H_{3,1}(f) - a_6 R_1 + a_5 R_2 - a_4 R_3,$$
(15)

where

$$H_{3,1}(f) = (a_3a_5 - a_4^2) - a_2(a_2a_5 - a_3a_4) + a_3(a_2a_4 - a_3^2)$$
(16)

and R_1 , R_2 , and R_3 are determinants of order 3, given by

$$R_1 = (a_3a_6 - a_4a_5) - a_2(a_2a_6 - a_3a_5) + a_4(a_2a_4 - a_3^2),$$
(17)

$$R_2 = (a_4 a_6 - a_5^2) - a_2 (a_3 a_6 - a_4 a)_5 + a_3 (a_3 a_5 - a_4^2), \quad (18)$$

$$R_3 = a_2 (a_4 a_6 - a_5^2) - a_3 (a_3 a_6 - a_4 a_5) + a_4 (a_3 a_5 - a_4^2).$$
(19)

Also,

$$T_{4,1}(f) = a_1 C_1 - a_2 C_2 + a_3 C_3 - a_4 C_4,$$
(20)

where

$$C_1 = a_1 \left(a_1^2 - a_2^2 \right) - a_4 \left(a_1 a_4 - a_2 a_3 \right) + a_3 \left(a_4 a_2 - a_1 a_3 \right),$$
(21)

$$C_{2} = a_{2} \left(a_{1}^{2} - a_{2}^{2}\right) - a_{4} \left(a_{1}a_{3} - a_{2}a_{4}\right) + a_{3} \left(a_{2}a_{3} - a_{1}a_{4}\right),$$

$$C_{3} = a_{2} \left(a_{1}a_{4} - a_{2}a_{3}\right) - a_{3} \left(a_{1}^{2} - a_{3}^{2}\right) + a_{4} \left(a_{1}a_{2} - a_{3}a_{4}\right),$$

$$C_{4} = a_{2} \left(a_{2}a_{4} - a_{1}a_{3}\right) - a_{3} \left(a_{1}a_{2} - a_{3}a_{4}\right) + a_{4} \left(a_{1}^{2} - a_{4}^{2}\right).$$
(22)

As, from (7), $H_{4,1}(f)$ is a polynomial of six coefficients of function f of the given class, these coefficients are taken as

 a_2 , a_3 , a_4 , a_5 , a_6 , and a_7 . However, there is a connection between these coefficients and the coefficients of function p in the class P in many problems. Consider that $f \in C_s$ has form (1); then, there is a Schwartz function w(z) with w(0) = 0 and |w(z)| < 1, such that

$$\frac{(zf'(z))'}{f'(z)} = 1 + \sin(w(z)).$$
(23)

Now,

$$\frac{(zf'(z))'}{f'(z)} = \frac{1 + \sum_{n=2}^{\infty} n^2 a_n z^{n-1}}{1 + \sum_{n=2}^{\infty} n a_n z^{n-1}}$$

= 1 + 2a_2z + (6a_3 - 4a_2^2)z^2
+ (-18a_2a_3 + 12a_4 + 8a_2^3)z^3
+ (-32a_2a_4 + 20a_5 - 18a_3^2 + 48a_3a_2^2 - 16a_2^4)z^4
+ (-50a_2a_5 + 30a_6 - 60a_4a_3
+ 80a_4a_2^2 + 90a_2a_3^2 - 120a_3a_2^3 + 32a_2^5)z^5 + \dots
(24)

Consider

$$p(z) = \frac{1+w(z)}{1-w(z)} = 1 + c_1 z + c_2 z^2 + c_3 z^3 + \cdots.$$
(25)

Since we have $p \in P$,

$$w(z) = \frac{p(z) - 1}{1 + p(z)} = \frac{c_1 z + c_2 z^2 + c_3 z^3 + \dots}{2 + c_1 z + c_2 z^2 + c_3 z^3 + \dots}.$$
 (26)

Also,

$$1 + \sin(w(z)) = 1 + \frac{1}{2}c_{1}z + \left(\frac{1}{2}c_{2} - \frac{1}{4}c_{1}^{2}\right)z^{2} + \left(\frac{1}{2}c_{3} - \frac{1}{2}c_{1}c_{2} + \frac{5}{48}c_{1}^{3}\right)z^{3} \\ + \left(\frac{5}{16}c_{2}c_{1}^{2} - \frac{1}{32}c_{1}^{4} + \frac{1}{2}c_{4} - \frac{1}{2}c_{1}c_{3} - \frac{1}{4}c_{2}^{2}\right)z^{4} \\ + \left(\frac{1}{3840}c_{1}^{5} + \frac{5}{16}c_{1}c_{2}^{2} - \frac{1}{8}c_{2}c_{1}^{3} - \frac{1}{2}c_{2}c_{3} + \frac{5}{16}c_{3}c_{1}^{2} \\ - \frac{1}{2}c_{1}c_{4} + \frac{1}{2}c_{5}\right)z^{5} + \cdots.$$

$$(27)$$

On comparing coefficients between (25) and (28), we get

$$a_{2} = \frac{1}{4}c_{1},$$

$$a_{3} = \frac{1}{12}c_{2},$$

$$a_{4} = -\frac{1}{96}c_{1}c_{2} - \frac{1}{576}c_{1}^{3} + \frac{1}{24}c_{3},$$

$$a_{5} = -\frac{1}{960}c_{2}c_{1}^{2} + \frac{1}{1152}c_{1}^{4} - \frac{1}{120}c_{1}c_{3} - \frac{1}{160}c_{2}^{2} + \frac{1}{40}c_{4},$$

$$a_{6} = -\frac{11}{28800}c_{1}^{5} + \frac{71}{34560}c_{2}c_{1}^{3} + \left(\frac{1}{1152}c_{2}^{2} - \frac{1}{160}c_{4}\right)c_{1} - \frac{7}{720}c_{2}c_{3} + \frac{1}{60}c_{5},$$
(28)

and

$$a_{7} = \frac{2399}{14515200}c_{1}^{6} - \frac{347}{241920}c_{2}c_{1}^{4} + \frac{1}{756}c_{3}c_{1}^{3} + \left(\frac{1}{3360}c_{4} + \frac{29}{16128}c_{2}^{2}\right)c_{1}^{2}$$

$$+ \left(-\frac{1}{210}c_{5} + \frac{23}{10080}c_{2}c_{3}\right)c_{1} + \frac{1}{84}c_{6} - \frac{1}{252}c_{3}^{2} + \frac{5}{8064}c_{2}^{3} - \frac{5}{672}c_{4}c_{2}.$$
(29)

$$H_{3,1}(f) = -\frac{19}{331776}c_{1}^{6} + \frac{1}{34560}c_{2}c_{1}^{4} + \frac{23}{34560}c_{3}c_{1}^{3} + \left(-\frac{11}{46080}c_{2}^{2} - \frac{1}{640}c_{4}\right)c_{1}^{2}$$
(20)

$$\begin{aligned} R_{2} &= \frac{1}{66355200}c_{1}^{2} + \frac{1}{12441600}c_{2}^{2}c_{1}^{2} + \frac{1}{2764800}c_{3}^{2}c_{1}^{2} + \left(\frac{1}{4147200}c_{2}^{2} - \frac{1}{23040}c_{4}^{2}\right)c_{1}^{2} \\ &+ \left(\frac{7}{64800}c_{2}c_{3} - \frac{1}{34560}c_{5}\right)c_{1}^{3} + \left(-\frac{1}{6400}c_{3}^{2} + \frac{7}{38400}c_{2}c_{4} - \frac{37}{921600}c_{2}^{3}\right)c_{1}^{2} \\ &+ \left(\frac{1}{5400}c_{2}^{2}c_{3} - \frac{1}{1920}c_{2}c_{5} + \frac{1}{2400}c_{3}c_{4}\right)c_{1} + \frac{1}{1440}c_{3}c_{5} + \frac{7}{14400}c_{2}^{2}c_{4} - \frac{1}{1600}c_{4}^{2} \\ &- \frac{19}{230400}c_{2}^{4} - \frac{19}{34560}c_{2}c_{2}^{3}, \end{aligned}$$
(32)

and

$$R_{3} = -\frac{83}{4777574400}c_{1}^{9} + \frac{317}{796262400}c_{2}c_{1}^{7} - \frac{49}{66355200}c_{3}c_{1}^{6}$$

$$+ \left(-\frac{1}{122880}c_{4}\right)$$

$$-\frac{169}{132710400}c_{2}^{2}c_{2}c_{1}^{5}\left(-\frac{1}{138240}c_{5} + \frac{209}{8294400}c_{2}c_{3}c_{3}\right)c_{1}^{4} + \frac{61}{2764800}c_{2}c_{4}c_{3}^{2} - \frac{23}{2764800}c_{3}^{2} - \frac{2993}{199065600}c_{2}^{3}c_{1}^{3}$$

$$+ \left(-\frac{1}{23040}c_{2}c_{5} + \frac{11}{5529600}c_{2}^{2}c_{3}c_{3}\right)$$

$$+ \frac{1}{25600}c_{3}c_{4}c_{1}^{2} + \left(\frac{1}{12800}c_{2}^{2}c_{4} - \frac{41}{8294400}c_{2}^{4}c_{4}^{2}\right)$$

$$+ \frac{1}{5760}c_{3}c_{5} - \frac{1}{6400}c_{4}^{2} - \frac{29}{276480}c_{2}c_{3}^{2}c_{3}c_{4} - \frac{1}{13824}c_{3}^{3} + \frac{1}{5760}c_{2}c_{3}c_{4}.$$
(33)

$$C_{1} = -\frac{1}{331776}c_{1}^{6} - \frac{1}{9216}c_{1}^{4}c_{2} + \frac{1}{6912}c_{1}^{3}c_{3} + \left(-\frac{1}{16} - \frac{5}{9216}c_{2}^{2}\right)c_{1}^{2} + \frac{1}{384}c_{1}c_{2}c_{3} + 1 - \frac{1}{576}c_{3}^{2} - \frac{1}{144}c_{2}^{2},$$
(34)

$$\begin{split} C_{2} &= \frac{1}{1327104}c_{1}^{7} + \frac{1}{110592}c_{1}^{5}c_{2} - \frac{1}{27648}c_{1}^{4}c_{3} + \left(\frac{1}{36864}c_{2}^{2} - \frac{1}{64}\right) \\ &+ \frac{1}{3456}c_{2}c_{1}^{3} - \frac{1}{4608}c_{1}^{2}c_{2}c_{3} + \left(\frac{1}{4} + \frac{1}{288}c_{2}^{2} + \frac{1}{2304}c_{3}^{2}\right)c_{1} \\ &- \frac{1}{144}c_{3}c_{2}, \end{split}$$

$$C_{3} = -\frac{1}{3981312}c_{1}^{6}c_{2} + \left(-\frac{1}{1152} - \frac{1}{331776}c_{2}^{2}\right)c_{1}^{4} + \frac{1}{82944}c_{1}^{3}c_{3}c_{2} + \left(-\frac{1}{110592}c_{2}^{3} - \frac{1}{96}c_{2}\right)c_{1}^{2} + \left(\frac{1}{48}c_{3} + \frac{1}{13824}c_{2}^{2}c_{3}\right)c_{1} + \frac{1}{1728}c_{2}^{3} - \frac{1}{6912}c_{3}^{2}c_{2} - \frac{1}{12}c_{2},$$
(36)

Similarly, in case of Toeplitz determinants,
$$\begin{split} C_4 &= \frac{1}{191102976} c_1^9 + \frac{1}{10616832} c_1^7 c_2 - \frac{1}{2654208} c_1^6 c_3 \\ &+ \left(\frac{1}{1769472} c_2^2 - \frac{1}{9216}\right) c_1^5 - \frac{1}{221184} c_1^4 c_2 c_3 \\ &+ \left(\frac{1}{884736} c_2^3 - \frac{1}{1536} c_2 \right) \\ &- \frac{1}{82944} c_2^2 + \frac{1}{110592} c_3^2 - \frac{1}{576} c_1^3 + \left(\frac{1}{384} c_3 - \frac{1}{73728} c_2^2 c_3\right) c_1^2 \\ &+ \left(-\frac{15}{96} c_2 + \frac{1}{18432} c_3^2 c_2 - \frac{1}{13824} c_2^3\right) c_1 + \frac{1}{24} c_3 - \frac{1}{13824} c_3^3 \\ &+ \frac{1}{3456} c_2^2 c_3. \end{split}$$

By using the previous computations, we prove the following.

Theorem 1. If the function $f \in C_s$ and is of form (1), then 112267159597

$$|H_{4,1}(f)| \le \frac{11220/159597}{15049359360000} \approx 0.0074599\cdots$$
 (38)

Proof. As $f \in C_s$, then by using (30)–(33) in (15), we get

$$\begin{split} H_{4,1}(f) &= \frac{38723}{481579499520000} c_1^{12} + \frac{23}{2322432} c_3^4 - \frac{1159}{6967296000} c_2^5 \\ &\quad - \frac{4169}{6967296000} c_1^7 c_5 - \frac{19}{27869184} c_9 c_1^6 \\ &\quad - \frac{19}{1451520} c_9 c_2^3 - \frac{1}{48384} c_9 c_3^2 - \frac{17}{6967296} c_3^2 c_3^3 + \frac{2539}{580608000} c_2^4 c_4 + \frac{1}{1792000} c_2^2 c_4^2 \\ &\quad + \frac{1}{57600} c_1^2 c_5^2 - \frac{1}{43200} c_2 c_5^2 - \frac{10}{64000} c_4^3 + \frac{1619}{580608000} c_3^1 c_3 c_2 c_4 - \frac{1}{67200} c_1^2 c_5 c_2 c_3 - \frac{1}{53760} c_1 c_5 c_2 c_4 \\ &\quad - \frac{713}{48384000} c_2^2 c_1 c_3 c_4 + \frac{11}{483840} c_6 c_2 c_1 c_3 + \frac{247}{580608000} c_2 c_1 c_3^3 + \frac{589}{145152000} c_5^2 c_5 c_2 \\ &\quad + \frac{53}{36864000} c_2 c_1^4 c_3^3 + \frac{23}{22903040} c_6 c_1^3 c_3 + \frac{563}{32256000} c_1^2 c_4^2 c_2 + \frac{1}{64000} c_1 c_4^2 c_3 + \frac{29}{1036800} c_2^2 c_3 c_5 \\ &\quad + \frac{1}{28800} c_5 c_5 c_4 - \frac{913}{348364800} c_1^3 c_5 c_2^2 - \frac{43}{2419200} c_1 c_5 c_3^2 - \frac{3019}{1032192000} c_2^2 c_5^2 c_3 \\ &\quad - \frac{11}{3870720} c_9 c_1^2 c_2^2 + \frac{67}{96768000} c_1 c_5 c_3^3 + \frac{1}{2903040} c_6 c_2 c_4 + \frac{151}{96768000} c_1^2 c_4 c_3^2 + \frac{251}{139345920} c_1^6 c_2 c_4 \\ &\quad + \frac{1073}{580608000} c_1^2 c_4 c_3^2 - \frac{11077}{55738368000} c_1^7 c_2 c_3 + \frac{1}{40320} c_6 c_2 c_4 - \frac{1}{3628800} c_1^4 c_5 c_3 \\ &\quad - \frac{1919}{1741824000} c_2^4 c_1 c_3 - \frac{29}{20736000} c_5^5 c_5 c_4 - \frac{1}{53760} c_6 c_1^2 c_4 - \frac{30703}{464864000} c_2^2 c_1^4 c_4 \\ &\quad + \frac{195313}{41803776000} c_1^3 c_5 c_3^2 + \frac{4699}{2322432000} c_1^2 c_2^2 c_3^2 - \frac{17}{2419200} c_1^3 c_5 c_4 - \frac{1}{37800} c_3^2 c_2 c_4 \\ &\quad - \frac{136291}{20065812480000} c_1^{10} c_2 + \frac{14201}{1003290624000} c_1^3 c_5^3 + \frac{852083}{668860416000} c_2^4 c_1^4 \\ &\quad - \frac{40073}{1003290624000} c_1^6 c_3^3 + \frac{8419}{289728000} c_1^6 c_3^3 + \frac{852083}{668860416000} c_2^4 c_1^4 \\ &\quad - \frac{5947}{1741824000} c_1^2 c_3^3 + \frac{1}{2867200} c_1^2 c_4^2 - \frac{391}{217728000} c_1^2 c_5^2. \end{split}$$

(39)

$$\begin{split} H_{4,1}(f) &= \frac{1}{481579499520000} \left\{ 38723c_1^{12} - 3270984c_1^{10}c_2 + 6816480c_1^9c_3 \\ &+ \left(24170040c_2^2 \right) \\ &- 4656960c_4 \right)c_1^8 + \left(-288161280c_5 - 95705280c_2c_3 \right)c_1^7 + \left(-328320000c_6 \right) \\ &+ 436440960c_2^2 - 19235040c_3^2 + 867456000c_4c_2 \right)c_1^6 + \left(-673505280c_3c_4 \right) \\ &- 1408544640c_3c_2^2 + 1954160640c_2c_5 \right)c_1^5 + \left(613499760c_2^4 + 165888000c_6c_2 \right) \\ &- 3183287040c_2^2c_4 + 167961600c_4^2 - 132710400c_3c_5 + 692375040c_2c_3^2 \right)c_1^4 \\ &+ \left(-1644226560c_3^3 + 1342863360c_2c_3c_4 + 2250005760c_2^3c_3 \right) \\ &+ 3815424000c_6c_3 - 3384115200c_5c_4 - 1262131200c_2^2c_5 \right)c_1^3 \\ &+ \left(8360755200c_5^2 - 8957952000c_6c_4 - 7166361600c_5c_2c_3 - 1368576000c_6c_2^2 \right) \\ &+ 8405544960c_4^2c_2 + 751472640c_4c_3^2 - 864829440c_2^5 + 974384640c_2^2c_3^2 \\ &+ 889989120c_4c_3^2 \right)c_1^2 + \left(3334348800c_5c_3^2 + 7524679680c_4^2c_3 \right) \\ &+ 10948608000c_6c_2c_3 + 2048716800c_2c_3^3 - 7096688640c_2^2c_3c_4 \\ &- 8957952000c_5c_2c_4 - 530565120c_2^4c_3 - 8559820800c_5c_3^2 \right)c_1 \\ &- 1175040000c_3^2c_2^3 + 2105948160c_2^4c_4 - 12740198400c_3^2c_2c_4 \\ &+ 11943936000c_6c_2c_4 + 16721510400c_5c_3c_4 - 7524679680c_4^3 + 4769280000c_3^4 \\ &- 80110080c_6^2 - 9953280000c_6c_3^2 - 11147673600c_2c_5^2 + 13470105600c_2^2c_3c_5 \\ &- 6303744000c_6c_3^2 + 268738560c_2^2c_4^2 \right]. \end{split}$$

After rearranging the terms, we get

$$\begin{split} H_{4,1}(f) &= \frac{1}{481579499520000} \\ &\cdot \left\{ -38723A_6 (c_2 - c_1^2)^3 - 2961200c_2A_5^2 + 6661588c_3A_3A_6 \\ &- 448912A_6 (c_3 - c_1c_2)^2 - 45407915c_4A_3A_5 - 288238726c_1 \\ &A_6 \Big(c_5 - \frac{1}{288238726}c_2c_3 \Big) - 328281275A_6 \Big(c_6 - \frac{403543209}{328281275}c_3^2 \Big) - A_6 (c_6 - 827940501c_2c_4) - A_6 \\ &\cdot (c_6 - 45022105c_2^3) + 519121748c_2A_5 \Big(c_5 - \frac{1592920565}{519121748}c_2c_3 \Big) + 635357352 \\ &\cdot c_3A_4 (c_5 - c_1c_4) - 1475634554c_2A_4 \Big(c_6 - \frac{2857592366}{1475634554}c_3^2 \Big) - 371563976c_5A_3 \\ &\cdot (c_4 - c_1c_3) + 158880018c_4^2A_3c_1 + 769228159c_2^4A_3c_1 + 1052977475c_2^2c_4A_3c_1 \\ &+ 5135210696 \Big(c_2 - c_1^2 \Big) \big(c_4 - c_1c_3 \Big) \Big(c_6 - \frac{3289911688}{5135210696}c_3^2 \Big) - 3872726611 \Big(c_2 - c_1^2 \Big) \\ &\cdot \Big(c_4 - \frac{2520764034}{3872726611}c_2^2 \Big) \big(c_6 - c_1c_5 \big) - 4015593821A_3c_2c_4c_3 - 4121711992A_3c_3c_3^2 \\ &- 8680311746c_4c_1^2 \Big(c_6 - \frac{11237134463}{8680311746}c_2c_4 \Big) - 24834316c_3^2c_1^2 \end{split}$$

$$\cdot \left(c_{4} - \frac{11757634360}{24834316}c_{2}^{2}\right) - 8937232652c_{5}\left(c_{2} - c_{1}^{2}\right)\left(c_{5} - \frac{7761341290}{8937232652}c_{2}c_{3}\right) - 1898862450c_{2}^{3}c_{1}^{2}\left(c_{4} - \frac{430044643}{1898862450}c_{2}^{2}\right) - 1305360709c_{2}^{2}c_{1}^{2}c_{6} + 17052974791 \cdot c_{3}c_{2}c_{1}\left(c_{6} - \frac{13683368690}{17052974791}c_{2}c_{4}\right) - 13635892999c_{5}c_{2}c_{1}\left(c_{4} - \frac{3937309300}{13635892999}c_{2}^{2}\right) + \left(-4485418959c_{2}c_{3}^{3} - 4542515027c_{2}^{4}c_{3} + 6721360728c_{4}^{2}c_{3}\right)c_{1} + 368323828c_{5}c_{1} \cdot \left(c_{6} - \frac{10733372524}{368323828}c_{3}^{2}\right) - 2726601496c_{2}c_{5}\left(c_{5} - \frac{7692726219}{2726601496}c_{2}c_{3}\right) + 10690320416 \cdot c_{2}c_{4}\left(c_{6} - \frac{4233634378}{10690320416}c_{3}^{2}\right) + 2843921897c_{2}^{4}\left(c_{4} - \frac{35126698}{2843921897}c_{2}^{2}\right) - 9215242838 \cdot c_{3}^{2}\left(c_{6} - \frac{4359524115}{9215242838}c_{3}^{2}\right) - 7632171375c_{6}c_{2}^{3} - 2720144201c_{3}^{2}c_{2}^{3} + 17732972519 \cdot c_{5}c_{3}c_{4} - 328281277c_{6}^{2} - 7524679680c_{4}^{3} - 31507775c_{2}c_{1}^{7}c_{3} - 1387142442c_{2}^{2}c_{4}^{2}.$$

$$(41)$$

After using triangular inequalities and lemmas, we get the following expression:

$$\begin{split} \left| H_{4,1}\left(f\right) \right| &= \frac{1}{481579499520000} \\ &\left\{ (38723 \times 2) \times 2^3 + 4(2961200 \times 2) + 4 \times (6661588 \times 2) + (448912 \times 2) \times 2^2 + 2 \times (45407915 \times 2) \times 2 + 2(288238726 \times 2) \times 2 + \frac{957610286}{328281275} \times (328281275 \times 2) + 2 \times 3311762002 + 2 \times (519121748 \times 2) \times \frac{1333359691}{129780437} + 2 \times (635357352 \times 2) \times 2 + 2 \times (1475634554 \times 2) \times \frac{4239550178}{737817277} + 371563976 \times (2) \times (2) \times (2) + 158880018 \times (4) \times (2) \times (2) + 769228159 \times (16) \times (2) \times (2) + 1052977475 \times (4) \times (2) \times (2) \times (2) + 5135210696 \times (2) \times (2) \times \frac{51593310}{91700191} + 872726611 \times (2) \times (2) \times (2) + 4015593821 \times (16) \end{split}$$
(42)
$$&+ 4121711992 \times (32) + 8680311746 \times (8) \times \frac{13793957180}{4340155873} + 24834316 \times (16) \times \frac{11745217202}{6208579} \\ &+ 8937232652 \times (2) \times (2) \times (2) + 1898862450 \times (32) \times (2) + 1305360709 \times (32) \\ &+ 17052974791 \times (8) \times (2) + +13635892999 \times (8) \times (2) + (4485418959) \\ &+ 6721360728 \times (8) \times (2) + 368323828 \times (4) \times \frac{10549210610}{92080957} + 2726601496 \times (4) \times \frac{6329425471}{681650374} \\ &+ 10690320416 \times (4) \times (2) + 2843921897 \times (16) \times (2) + 9215242838 \times (4) \times (2) \\ &+ 7632171375 \times (16) + 2720144201 \times (32) + 17732972519 \times (8) + 328281277 \times (4) \\ &+ 7524679680 \times (8) + 31507775 \times (512) + 1387142442 \times (16)]. \end{split}$$

Hence,

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$$H_{4,1}(f) \Big| \le \frac{3592549107104}{481579499520000} \approx 0.0074599, \tag{43}$$

which completes the proof.

Theorem 2. If the function $f \in C_s$ and is of form (1), then

 $\left|T_{4,1}\left(f\right)\right| \le \frac{501434459}{286654464} \approx 1.7493.$ (44)

Proof. As $f \in C_s$, using (34)–(37) in (20), we get

$$\begin{split} T_{4,1}(f) &= 1 + \frac{1}{256}c_1^4 - \frac{1}{288}c_3^2 - \frac{1}{165888}c_1^6 - \frac{1}{8}c_1^2 - \frac{1}{72}c_2^2 + \frac{5}{576}c_1c_2c_3 + \frac{1}{9216}c_1^3c_3c_2 \\ &+ \frac{1}{3456}c_1^3c_3 - \frac{1}{221184}c_1^6c_2 + \frac{1}{497664}c_1^3c_3c_2^2 - \frac{1}{5308416}c_1^3c_3^3c_3 \\ &+ \frac{1}{82944}c_1c_2^3c_3 - \frac{1}{331776}c_1c_2c_3^3 + \frac{1}{884736}c_1^2c_2^2c_3^2 - \frac{1}{10616832}c_1^5c_2^2c_3 \\ &- \frac{1}{63700992}c_1^7c_2c_3 + \frac{1}{2654208}c_1^4c_2c_3^2 - \frac{13}{4608}c_1^2c_2^2 + \frac{1}{55296}c_1^5c_3 \\ &- \frac{1}{73728}c_1^4c_2^2 - \frac{1}{4608}c_1^2c_3^2 - \frac{1}{23887872}c_1^6c_2^2 - \frac{1}{1990656}c_1^4c_2^3 \\ &- \frac{1}{663552}c_1^2c_2^4 - \frac{1}{41472}c_2^2c_3^2 + \frac{1}{84934656}c_1^4c_2^4 + \frac{1}{127401984}c_1^6c_2^3 \\ &+ \frac{1}{509607936}c_1^8c_2^2 + \frac{1}{4586471424}c_1^{10}c_2 - \frac{1}{1146617856}c_1^9c_3 + \frac{1}{31850496}c_1^6c_3^2 \\ &- \frac{1}{1990656}c_1^3c_3^3 - \frac{1}{2654208}c_1^8 + \frac{1}{20736}c_2^4 + \frac{1}{110075314176}c_1^{12}. \end{split}$$

 \Box

Rearranging the terms, we may write

$$\begin{split} T_{4,1}(f) &= \frac{1}{110075314176} \Big\{ c_1^{12} + 24c_1^{10}c_2 - 96c_1^9c_3 + \Big(-41472 + 216c_2^2\Big)c_1^8 - 1728c_1^7c_2c_3 \\ &+ \Big(-663552 + 3456c_3^2 - 497664c_2 + 864c_2^3 - 4608c_2^2\Big)c_1^6 \\ &+ \Big(-10368c_2^2c_3 + 1990656c_3\Big)c_1^5 \\ &+ \Big(-1492992c_2^2 + 41472c_3^2c_2 - 55296c_2^3 - 39813120c_2 + 1296c_2^4 + 429981696\Big)c_1^4 \\ &+ \Big(11943936c_2c_3 + 31850496c_3 - 55296c_3^3 - 20736c_2^3c_3 + 221184c_3c_2^2\Big)c_1^3 \\ &+ \Big(-310542336c_2^2 - 13759414272 + 124416c_2^2c_3^2 - 165888c_2^4 - 23887872c_3^2\Big)c_1^2 \\ &+ \Big(1327104c_2^3c_3 + 955514880c_2c_3 - 331776c_2c_3^3\Big)c_1 \\ &- 1528823808c_2^2 - 382205952c_3^2 + 331776c_3^4 + 5308416c_2^4 \\ &+ 110075314176 - 2654208c_2^2c_3^2\Big\}. \end{split}$$

After rearranging the terms, we get

$$\begin{split} T_{41}(f) &= \frac{1}{110075314176} \left\{ -A_{6}(c_{2}-c_{1}^{2})^{3} - 100A_{3}A_{6}(c_{3}-\frac{32}{100}c_{1}c_{2}) - 2c_{1}A_{6}(c_{3}-\frac{3}{2}c_{4}) \right. \\ &\quad - 2570c_{1}c_{2}A_{6}(c_{3}-\frac{416}{2570}c_{1}c_{2}) - 41472A_{3}A_{5} + 2156544A_{5}(c_{3}-\frac{746496}{2156544}c_{1}c_{2}) \\ &\quad - 22704c_{2}^{2}A_{3}(c_{5}-\frac{2467}{22704}c_{1}c_{2}) - 312c_{4}A_{5}(c_{3}-\frac{100}{312}c_{1}c_{2}) + 29c_{2}A_{4}(c_{6}-\frac{68}{29}c_{5}c_{5}) \\ &\quad + (-663552+c_{6}-4608c_{2}^{2}+3955c_{3}^{2})A_{6}+22311936c_{2}A_{3}(c_{3}-\frac{4022784}{22311936}c_{4}c_{2}) \\ &\quad - 902c_{2}^{2}A_{3}(c_{5}-\frac{1389}{902}c_{1}c_{4}) - 12A_{3}c_{4}(c_{5}-\frac{9}{12}c_{1}c_{4}) - 104c_{3}A_{5}(c_{6}-\frac{208}{104}c_{1}c_{5}) \\ &\quad - 70616c_{3}^{2}A_{3}(c_{3}-\frac{69470}{70616}c_{1}c_{2}) - 102891c_{3}^{3}A_{5}(c_{3}-\frac{9058}{102891}c_{2}c_{2}) + 239616c_{2}^{2}A_{3} \\ &\quad (c_{7}-\frac{78336}{239616}c_{1}c_{2}) + 41472A_{3}(c_{5}-\frac{82944}{41472}c_{4}) + 12498c_{3}c_{4}c_{1}^{2}(c_{5}-\frac{8344}{12498}c_{1}c_{2}) \\ &\quad + (5c_{6}c_{2}-4313088c_{2}+429981696)A_{4}+34504704c_{3}A_{5}-633c_{4}^{2}c_{1}(c_{5}-\frac{212}{633}c_{1}c_{2}) \\ &\quad + 4313088c_{3}c_{1}(c_{4}-\frac{30233088}{4313088}c_{1}c_{3}) - 188c_{5}c_{5}c_{1}(c_{4}-\frac{5216}{188}c_{5}c_{3}) + 1857024 \\ &\quad c_{3}^{2}c_{1}(c_{3}-\frac{294912}{1857024}c_{1}c_{2}) - 144138c_{2}^{4}c_{1}(c_{3}-\frac{11076}{144138}c_{5}c_{2}) - 859963392c_{1}. \\ &\quad (c_{3}-\frac{1289945088}{859963392}c_{1}c_{2}) + 4097433c_{2}^{2}c_{1}(c_{5}-\frac{63592}{4204735}c_{1}c_{2}) - 859963392c_{1}. \\ &\quad (c_{3}-\frac{6702}{21698}c_{5}c_{2}) + 40974336c_{2}^{2}c_{1}(c_{3}-\frac{435953664}{106490735}c_{1}c_{2}) \\ &\quad + 746496c_{5}c_{1}(c_{5}-\frac{14229292}{746496}c_{5}c_{1}) + 9216c_{2}^{2}c_{1}(c_{5}-\frac{13827}{2216}c_{5}c_{1}) - 2608c_{6}c_{2}c_{1} \\ &\quad (c_{5}-\frac{6710}{2008}c_{5}c_{2}) + 1106804736c_{2}c_{1}(c_{5}-\frac{435953664}{106804735}c_{1}c_{2}) \\ &\quad + (4c_{3}^{2}-1990656c_{4}-13759414272)c_{1}^{2}+23260c_{2}^{2}c_{5}(c_{5}-\frac{2136}{23606}c_{5}c_{5}) \\ &\quad + (4c_{3}^{2}-1990656c_{4}-13759414272)c_{1}^{2}+23260c_{2}^{2}c_{5}) \\ &\quad + (4c_{3}^{2}-1990656c_{6}-13759414272)c_{1}^{2}+232060c_{2}^{2}c_{5})$$

After using the triangular inequality and above-stated lemmas, we get

$$\begin{split} |T_{4,1}(f)| &= \frac{1}{110075314176} \\ & \left\{ 2 \times (2)^3 + 100 \times (2) \times (2) \times (2) + 2 \times (2) \times (2) \times (2) + 2570 \times (2) \times (2) \times (2) \right. \\ & (2) \times (2) + 41472 \times (2) \times (2) + 2156544 \times (2) \times (2) + 22704 \times (4) \times (2) \times (2) + 312 \times (2) \times (2) \times (2) + 29 \times (2) \times (2) \times (2) + 29 \times (2) \times (2) + 29 \times (2) \times (2) \times (2) + 29 \times (2) + 29 \times (2) \times (2) \times (2) \times (2) + 102891 \times (8) \times (2) \times (2) \times (2) + 104 \times (2) \times (2) \times (2) \times (2) + 102891 \times (8) \times (2) \times (2) + 2 \times (239616 \times 4) \times 2 + 6 \times (41472 \times 2) + 2 \times (2 \times (12498 \times 2) \times 4) + (5 \times 4 + 43130880 \times 2 + 429981696) \times 2 + 34504704 \times (2) \times (2) + 2 \times (633 \times 4) \times 2 + 2 \times (4313088 \times 2) \times \frac{677}{26} + \frac{5122}{47} \times (2 \times (188 \times 2) \times 2) + 2 \times (4 \times (517010 \times 2) \times 2) + 2 \times (746496 \times 2) \times 6 + 2 \times (9216 \times 4) \times 4 + 2 \times (2 \times (2608 \times 2)) \times (2) + 2 \times (40974336 \times 4) \times 2 + 2 \times (2 \times (40985 \times 4) \times 2) + 2 \times ((1106804736 \times 2) \times 2) + (4 \times 4 + 1990656 \times 2 + 13759414272) \times 4 + 23606 \times (4) \times (2) \times (2) + 3258 \times (4) \times (6) + 41803776 \times (2) \times \frac{130}{63} + 429981696 \times \frac{146}{9} + 9216 \times (8) \times (2) + 32 \times (2) \times (2) \times (2) + 2 \times (16) \times (10 \times 101075314176 + 416047104 \times (4) + 324 \times (2) \times (2) \times (2) + (4) + 18081792 \times (2) \times (4) + 2889216 \times (4) \times (4) + 2156544 \times (2) \times (2) + (2) + (4) + 18081792 \times (2) \times (4) + 2889216 \times (4) \times (4) + 2156544 \times (2) \times (2) + (2) + (4) + 18081792 \times (2) \times (4) + 2889216 \times (4) \times (4) + 2156544 \times (2) \times (2) + (2) + (4) + 18081792 \times (2) \times (4) + 2889216 \times (4) \times (4) + 2156544 \times (2) \times (2) +$$

This reduces to

$$\left|T_{4,1}\left(f\right)\right| \leq \frac{192550832256}{110075314176} = \frac{501434459}{286654464} \approx 1.7493\cdots, \quad (50)$$

which completes the proof.

2.2. Bounds of $|H_{4,1}|$ for the Set $C_s^{(3)}$. Let $n \in \mathbb{N} = \{1, 2, 3, ...\}$. Rotation of a domain \mathcal{D} about the origin through an angle of $2\pi/n$ containing \mathcal{D} onto itself is said to be *n*-fold symmetric. An analytic function f is *n*-fold symmetric in D if

$$f\left(e^{2\pi i/n}z\right) = e^{2\pi i/n}f\left(z\right) \tag{51}$$

holds for any $z \in D$. Denote $S^{(n)}$ as the set of *n*-fold univalent functions which have the following Taylor series form:

$$f(z) = z + \sum_{k=1}^{\infty} a_{nk+1} z^{nk+1} (z \in D).$$
 (52)

Denote $C^{(n)}$ as the subfamily of $S^{(n)}$ of *n*-fold symmetric convex functions. We can see that an analytic function *f* of form (52) belongs to the family $C^{(n)}$, if and only if

$$\frac{(zf'(z))'}{f'(z)} = p(z),$$
(53)

where $p \in P^{(n)}$. The family $P^{(n)}$ is defined as

$$P^{(n)} = \left\{ p \in P: \ p(z) = 1 + \sum_{k=1}^{\infty} c_{nk} z^{nk}, \ (z \in D) \right\}.$$
(54)

Now, consider the following.

Theorem 3. Let $f \in C_s^{(3)}$ be of form (52). Then,

$$H_{4,1}(f) \Big| \le \frac{1}{6048} \approx 0.00016534.$$
 (55)

Proof. Let $f \in C_s^{(3)}$ of form (52). Consider the function $p \in P^{(3)}$ as

$$p(z) = \frac{1 + w(z)}{1 - w(z)} = 1 + c_3 z^3 + c_6 z^6 + c_9 z^9 + \dots$$
 (56)

Now,

$$w(z) = \frac{p(z) - 1}{p(z) + 1}.$$
(57)

The class $C_s^{(3)}$ which is associated with the sine functions can be written in the following form:

$$\frac{(zf'(z))}{f'(z)} = 1 + \sin(w(z)), \ (z \in D).$$
(58)

By expanding and equating them, we get the following expression:

$$1 + 12a_4z^3 + \left(42a_7 - 48a_4^2\right)z^6 + \dots = 1 + \frac{1}{2}c_3z^3 + \left(\frac{1}{2}c_6 - \frac{1}{4}c_3^2\right)z^6 + \dots$$
(59)

This implies

$$a_4 = \frac{1}{24}c_3,$$

$$a_7 = -\frac{1}{252}c_3^2 + \frac{1}{84}c_6.$$
(60)

By using these coefficients, we can get $H_{3,1}(f)$, R_1, R_2 , and R_3 as

$$H_{3,1} = -\frac{1}{576}c_3^2,$$

$$R_1 = 0, R_2 = 0,$$

$$R_3 = -\frac{1}{13824}c_3^3.$$
(61)

By using these values in $H_{4,1}(f)$, we get

$$H_{4,1}(f) = a_7 H_{3,1}(f) - a_6 R_1 + a_5 R_2 - a_4 R_3$$

$$= \left(-\frac{1}{252}c_3^2 + \frac{1}{84}c_6\right)\left(-\frac{1}{576}c_3^2\right) - \left(\frac{1}{24}c_3\right)\left(-\frac{1}{13824}c_3^3\right)$$
$$= \frac{23}{2322432}c_3^4 - \frac{1}{48384}c_3^2c_6$$
$$= -\frac{1}{48384}c_3^2\left(c_6 - \frac{23}{48}c_3^2\right).$$
(62)

The triangle inequality and the application of Lemma 1 lead us to

$$\left|H_{4,1}(f)\right| = \frac{1}{6048} \approx 0.00016534,\tag{63}$$

which completes the proof.

3. Conclusion

In this paper, we have found the upper bounds of fourthorder Hankel and Toeplitz determinants, followed by a review of such findings obtained so far for certain analytic functions. We have studied them for the convex functions associated with the function $1 + \sin z$. A similar bound of the fourth-order Hankel determinant for 3-fold symmetric convex functions associated with $1 + \sin z$ has also been investigated.

Data Availability

No data were used to support this study.

Conflicts of Interest

The authors declare that there are no conflicts of interest regarding the publication of this article.

Authors' Contributions

All authors contributed equally to this study and approved the final manuscript.

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Research Article

Novel Analysis of Fractional-Order Fifth-Order Korteweg–de Vries Equations

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In this paper, the ρ -homotopy perturbation transformation method was applied to analysis of fifth-order nonlinear fractional Korteweg–de Vries (KdV) equations. This technique is the mixture form of the ρ -Laplace transformation with the homotopy perturbation method. The purpose of this study is to demonstrate the validity and efficiency of this method. Furthermore, it is demonstrated that the fractional and integer-order solutions close in on the exact result. The suggested technique was effectively utilized and was accurate and simple to use for a number of related engineering and science models.

1. Introduction

A number of researchers have recently become interested in fractional calculus, which was first developed during Newton's period. Within the fractional calculus structure, many interesting and significant steps have been discovered within the last thirty decades. A fractional derivative was invented as a result of the complexity of a heterogeneous phenomenon. The fractional derivative operators, by incorporating diffusion methods, are capable of capturing the attitudes of multidimensional media [1–4]. The use of differential equations of any scale has proved useful in showing a number of problems more quickly and accurately. Increasingly, scholars turned to generalized calculus to convey their viewpoints while analyzing complex phenomena in the context of mathematical methods using software [5–10]. Nonlinear impacts occur in several implemented scientific fields, such as fluid, mathematical biology, nonlinear image sensors, quantum field theory, kinetics, thermodynamics, and fluid dynamics. It is based on nonlinear partial differential equations of various degrees of complexity to model these processes. Partial differential equations are generally applied in the description of physical processes [11–15]. Most of the essential physical systems do not exhibit linear behavior. There is no way to determine the exact result of such nonlinear phenomena. Only techniques that are appropriate for solving nonlinear equations can be used to investigate this phenomenon [16–22].

In 1895, Korteweg and de Vries proposed a KdV equation to design Russell's soliton phenomenon, such as small and huge water waves. Solitons are steady solitary waves, which mean that these solitary waves are a particle. KdV equations are applied in different applied fields such as quantam mechanics, fluid dynamics, optics, and plasma physics. Fifth-order KdV form equations were utilized to analyze many nonlinear phenomena in particle physics [23–25]. It plays a vital role in the distribution of waves [26]. In their analysis, the KdV form equation has dispersive terms of the third and fifth-order relevant to the magnetoacoustic wave problem in cold plasma free collision plasma and dispersive terms appear near-critical angle propagation [27]. Plasma is a dynamic, quasineutral, and electrically conductive fluid. It consists of neutral particles, electrons, and ions. It consists of magnetic and electric areas due to the electrically conducting behavior of plasma. The mixture of particles and areas supports plasma waves of various forms. A magnetic lock is a less longitudinal ion dispersion. The magnetoacoustic wave behaves as an ionacoustic wave in the low magnetic field range, while in the low-temperature capacity, it acts as an Alfven wave [28, 29].

The general model for the analysis of magnetic propertiesacoustic waves in plasma and shallow water waves with surface tension is equated with the fifth order of KdV. Recent study reveals that the solutions to this equation for travelling waves do not vanish at infinity [30, 31]. Consider the well-known three types of the fifth-order KdV equations as follows [32, 33]: $D_{\tau}^{\gamma} \mathcal{V} + \mathcal{V}_{\zeta} + \mathcal{V}^{2} \mathcal{V}_{2\zeta} + \mathcal{V}_{\zeta} \mathcal{V}_{2\zeta} - 20 \mathcal{V}^{2} \mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \leq 1,$ (1)

with initial condition $\mathcal{V}(\zeta, 0) = 1/\zeta$,

$$D^{\beta}_{\tau}\mathcal{V} + \mathcal{V}\mathcal{V}_{\zeta} - \mathcal{V}\mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \le 1, \qquad (2)$$

with initial condition $\mathcal{V}(\zeta, 0) = e^{\zeta}$, and

$$D^{\beta}_{\tau}\mathcal{V} + \mathcal{V}\mathcal{V}_{\zeta} + \mathcal{V}_{3\zeta} - \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \le 1,$$
(3)

with initial condition $\mathcal{V}(\zeta, 0) = 105/169 \operatorname{sech}^4(\zeta - \phi/2\sqrt{13})$.

(1) and (2) are called fifth-order KdV equations and (3) is called the Kawahara equation. Analytic techniques for these mathematical model are particularly difficult to come across due to their severe nonlinearity. Several researchers have employed various analytical and computational strategies to the solution of linear and nonlinear KdV equations throughout the last decade, such as the multisymplectic method [34], variational iteration method [33], He's homotopy perturbation method [35], and Exp-function method [36].

Recently, Fahd and Abdeljawad [37] developed the Laplace transform of the generalized fractional Caputo derivatives. We established a novel methodology with ρ-Laplace transform for solving fractional differential equations with a generalized fractional Caputo derivative. The homotopy perturbation method is merged with the Laplace transform method to create a highly effective method for handling nonlinear terms which is known as the homotopy perturbation transformation technique. This technique can provide the result in quick convergent series. Ghorbani pioneered the use of He's polynomials in nonlinear terms [38-40]. Later on, many scholars utilized the homotopy perturbation transformation method for linear and nonlinear differential equations such as heat-like equations [41], Navier-Stokes equations [42], hyperbolic equation and Fisher's equation [43], and gas dynamic equation [44].

2. Basic Definitions

2.1. Definition. The fractional generalized integral of order β of a continuous function (CF) $g: [0, +\infty] \longrightarrow R$ is defined as [37]

$$\left(I^{\beta,\rho}g\right)(\zeta) = \frac{1}{\Gamma(\beta)} \int_0^{\zeta} \left(\frac{\zeta^{\rho} - s^{\rho}}{\rho}\right)^{\beta-1} \frac{g(s)ds}{s^{1-\rho}}, \quad \rho > 0, \ \zeta > 0, \ 0 < \beta < 1.$$
(4)

2.2. Definition. The order β fractional generalized derivative of a CF g: $[0, +\infty] \longrightarrow R$ is given as [37]

$$(D^{\beta,\rho}g)(\zeta) = (I^{1-\beta,\rho}g)(\zeta)$$
$$= \frac{1}{\Gamma(1-\beta)} \left(\frac{d}{d\zeta}\right) \int_0^{\zeta} \left(\frac{\zeta^{\rho} - s^{\rho}}{\rho}\right)^{-\beta-1} \frac{g(s)ds}{s^{1-\rho}}, \quad (5)$$
$$\rho > 0, \ \zeta > 0 \text{ and } 0 < \beta < 1.$$

2.3. Definition. The Caputo derivative of fractional-order β of a CF g: $[0, +\infty] \longrightarrow R$ is defined as [37]

$$\left(D^{\beta,\rho}g\right)(\zeta) = \frac{1}{\Gamma(1-\beta)} \left(\frac{d}{d\zeta}\right) \int_0^{\zeta} \left(\frac{\zeta^{\rho} - s^{\rho}}{\rho}\right)^{-\beta-1} \beta^n \frac{g(s)ds}{s^{1-\rho}},\tag{6}$$

where $\rho > 0$, $\zeta > 0$, $\beta = \zeta^{1-\beta} d/d\zeta$, and $0 < \beta < 1$.

2.4. Definition. The ρ -Laplace transform of a CF $g: [0, +\infty] \longrightarrow R$ is defined as [37]

$$L_{\rho}\left\{g\left(\zeta\right)\right\}(s) = \int_{0}^{\infty} e^{-s\zeta^{\rho}/\rho} g\left(\zeta\right) \frac{d\zeta}{\zeta^{1-\rho}}.$$
(7)

The fractional generalized Caputo derivative of ρ -Laplace transformation of a CF *g* is given by [37]

$$L_{\rho}\left\{D^{\beta,\rho}g(\zeta)\right\}(s) = s^{\beta}L_{\rho}\left\{g(\zeta)\right\} - \sum_{k=0}^{n-1} s^{\beta-k-1} \left(I^{\beta,\rho}\beta^{n}g\right)(0).$$
(8)

2.5. Definition. The generalized Mittag-Leffler function is defined by

$$E_{\beta,\rho}(z) = \sum_{k=0}^{\infty} \frac{z^{\beta}}{\Gamma(\beta k + \gamma)},$$
(9)

where $\beta > 0$, $\gamma > 0$, and $E_{\beta}(z) = E_{\beta,1}(z)$.

3. The Rod Map of the Proposed Method

Consider the general partial differential equation given as

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$$D_{\tau}^{\gamma} \mathcal{V}(\zeta, \tau) + M \mathcal{V}(\zeta, \tau) + N \mathcal{V}(\zeta, \tau) = h(\zeta, \tau), \quad \tau > 0, \ 0 < \gamma \le 1,$$
$$\mathcal{V}(\zeta, 0) = g(\zeta), \quad \nu \in \mathfrak{R}.$$
(10)

Applying ρ -Laplace transformation of (10), we get

 $L_{\rho}\left[D_{\tau}^{\gamma}\mathcal{V}(\zeta,\tau) + M\mathcal{V}(\zeta,\tau) + N\mathcal{V}(\zeta,\tau)\right] = L_{\rho}\left[h(\zeta,\tau)\right], \quad \tau > 0, \ 0 < \gamma \le 1,$ $\mu(\zeta,\tau) = \frac{1}{s}g(\zeta) + \frac{1}{s^{\beta}}L_{\rho}\left[h(\zeta,\tau)\right] - \frac{1}{s^{\beta}}L_{\rho}\left[M\mathcal{V}(\zeta,\tau) + N\mathcal{V}(\zeta,\tau)\right].$ (11)

Now, applying the inverse ρ -Laplace transform, we get

$$\mathcal{V}(\zeta,\tau) = F(\zeta,\tau) - L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \{ M \mathcal{V}(\zeta,\tau) + N \mathcal{V}(\zeta,\tau) \} \right],$$
(12)

where

$$F(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s} g(\zeta) + \frac{1}{s^{\beta}} L_{\rho} [h(\zeta, \tau)] \right]$$

$$= g(\nu) + L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} [h(\zeta, \tau)] \right].$$
(13)

Now, the perturbation procedure in terms of power series with parameter p is presented as

$$\mathscr{V}(\zeta,\tau) = \sum_{\kappa=0}^{\infty} p^{\kappa} \mathscr{V}_{\kappa}(\zeta,\tau), \tag{14}$$

where p is the perturbation parameter and $p \in [0, 1]$. The nonlinear term can be defined as

$$N\mathscr{V}(\zeta,\tau) = \sum_{\kappa=0}^{\infty} p^{\kappa} H_{\kappa}(\mathscr{V}_{\kappa}), \qquad (15)$$

where H_n are He's polynomials in terms of $\mathcal{V}_0, \mathcal{V}_1, \mathcal{V}_2, \dots, \mathcal{V}_n$ and can be calculated as

$$H_n(\mathscr{V}_0, \mathscr{V}_1, \dots, \mathscr{V}_n) = \frac{1}{\gamma(n+1)} D_p^{\kappa} \left[N\left(\sum_{\kappa=0}^{\infty} p^{\kappa} \mathscr{V}_{\kappa}\right) \right]_{p=0},$$
(16)

where $D_p^{\kappa} = \partial^{\kappa} / \partial p^{\kappa}$. Substituting (15) and (16) in (12), we get

 $\sum_{\kappa=0}^{\infty} p^{\kappa} \mathcal{V}_{\kappa}(\zeta,\tau) = F(\zeta,\tau) - p \times \left[L_{\rho}^{-1} \left\{ \frac{1}{s^{\beta}} L_{\rho} \left\{ M \sum_{\kappa=0}^{\infty} p^{\kappa} \mathcal{V}_{\kappa}(\zeta,\tau) + \sum_{\kappa=0}^{\infty} p^{\kappa} H_{\kappa}(\mathcal{V}_{\kappa}) \right\} \right\} \right].$ (17)

The coefficients comparison on both sides of p, we have $p^0: \mathcal{V}_0(\zeta, \tau) = F(\zeta, \tau),$

$$\begin{split} p^{1} \colon \mathcal{V}_{1}(\zeta,\tau) &= L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left(M \mathcal{V}_{0}(\zeta,\tau) + H_{0}(\mathcal{V}) \right) \right], \\ p^{2} \colon \mathcal{V}_{2}(\zeta,\tau) &= L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left(M \mathcal{V}_{1}(\zeta,\tau) + H_{1}(\mathcal{V}) \right) \right], \\ &\vdots \end{split}$$

$$\begin{split} p^{\kappa} &: \mathcal{V}_{\kappa}(\zeta,\tau) = L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \big(M \mathcal{V}_{\kappa-1}(\zeta,\tau) + H_{\kappa-1}(\mathcal{V}) \big) \bigg], \\ & \kappa > 0, \; \kappa \in N. \end{split}$$

The $\mathscr{V}\kappa(\zeta,\tau)$ component can be determined easily which quickly leads us to the convergent series. We can get $p \longrightarrow 1$:

$$\mathscr{V}(\zeta,\tau) = \lim_{M \longrightarrow \infty} \sum_{\kappa=1}^{M} \mathscr{V}_{\kappa}(\zeta,\tau).$$
(19)

4. Numerical Implementations

Example 1. Consider the fifth-order nonlinear KdV equation

$$D^{\beta}_{\tau}\mathcal{V} + \mathcal{V}_{\zeta} + \mathcal{V}^{2}\mathcal{V}_{2\zeta} - \mathcal{V}_{\zeta}\mathcal{V}_{2\zeta} - 20\mathcal{V}^{2}\mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \le 1,$$
(20)

with the IC

(18)

$$\mathscr{V}(\zeta,\tau) = \frac{1}{\zeta}.$$
(21)

Applying the ρ -Laplace transform on (20), we get

$$L_{\rho}\mathcal{V}(\zeta,\tau)] = \frac{1}{s\zeta} - \frac{1}{s^{\beta}} L_{\rho}\mathcal{V}_{\zeta} + \mathcal{V}^{2}\mathcal{V}_{2\zeta} + \mathcal{V}_{\zeta}\mathcal{V}_{2\zeta} - 20\mathcal{V}^{2}\mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta}\Big].$$
(22)

Next, using the inverse of ρ -Laplace transform of (22),

 $\left[\mathcal{V}(\zeta,\tau)\right] = \frac{1}{\zeta} - L_{\rho}^{-1} \left[\frac{1}{\varsigma^{\beta}} L_{\rho} \left[\mathcal{V}_{\zeta} + \mathcal{V}^{2} \mathcal{V}_{2\zeta} + \mathcal{V}_{\zeta} \mathcal{V}_{2\zeta} - 20 \mathcal{V}^{2} \mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta} \right] \right].$ (23)

Now, we apply HPM

$$\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) = \frac{1}{\zeta} - p \left[L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left[\left(\sum_{n=0}^{\infty} p^{n} H_{n}(\mathcal{V}) \right) + \left(\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) \right)_{\zeta} + \left(\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) \right)_{5\zeta} \right] \right] \right], \tag{24}$$

where $H_n(x)$ represents the nonlinear function of He's polynomial. For the first few components, we present He's polynomials

$$\begin{split} H_{0}(\mathcal{V}) &= \mathcal{V}_{0}^{2} (\mathcal{V}_{0})_{2\zeta} + (\mathcal{V}_{0})_{\zeta} (\mathcal{V}_{0})_{2\zeta} - 20\mathcal{V}_{0}^{2} (\mathcal{V}_{0})_{3\zeta}, \\ H_{1}(\mathcal{V}) &= \mathcal{V}_{0}^{2} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{0})_{2\zeta} + (\mathcal{V}_{0})_{\zeta} (\mathcal{V}_{1})_{2\zeta} + (\mathcal{V}_{0})_{2\zeta} (\mathcal{V}_{1})_{\zeta} - 20\mathcal{V}_{0}^{2} (\mathcal{V}_{1})_{3\zeta} - 40\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{0})_{3\zeta}, \\ H_{2}(\mathcal{V}) &= \mathcal{V}_{0}^{2} (\mathcal{V}_{2})_{2\zeta} + 2\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{0} \mathcal{V}_{2} (\mathcal{V}_{0})_{2\zeta} + \mathcal{V}_{1}^{2} (\mathcal{V}_{0})_{2\zeta} + (\mathcal{V}_{0})_{\zeta} (\mathcal{V}_{2})_{2\zeta} + (\mathcal{V}_{1})_{\zeta} (\mathcal{V}_{1})_{2\zeta} \\ &\quad + (\mathcal{V}_{0})_{2\zeta} (\mathcal{V}_{2})_{\zeta} - 20\mathcal{V}_{0}^{2} (\mathcal{V}_{2})_{3\zeta} - 40\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{1})_{3\zeta} - 40\mathcal{V}_{0} \mathcal{V}_{2} (\mathcal{V}_{0})_{3\zeta} - 20\mathcal{V}_{1}^{2} (\mathcal{V}_{0})_{3\zeta}, \\ H_{3}(\mathcal{V}) &= \mathcal{V}_{0}^{2} (\mathcal{V}_{3})_{2\zeta} + 2\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{2})_{2\zeta} + (\mathcal{V}_{1})_{2\zeta} (\mathcal{V}_{2})_{\zeta} + (\mathcal{V}_{0})_{2\zeta} (\mathcal{V}_{3})_{\zeta} - 20\mathcal{V}_{0}^{2} (\mathcal{V}_{3})_{3\zeta} - 40\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{2})_{2\zeta} \\ &\quad + (\mathcal{V}_{0})_{\zeta} (\mathcal{V}_{3})_{2\zeta} + (\mathcal{V}_{1})_{\zeta} (\mathcal{V}_{2})_{2\zeta} + (\mathcal{V}_{1})_{2\zeta} (\mathcal{V}_{2})_{\zeta} + (\mathcal{V}_{0})_{2\zeta} (\mathcal{V}_{3})_{\zeta} - 20\mathcal{V}_{0}^{2} (\mathcal{V}_{3})_{3\zeta} - 40\mathcal{V}_{0} \mathcal{V}_{1} (\mathcal{V}_{2})_{3\zeta} \\ &\quad - 40\mathcal{V}_{0}\mathcal{V}_{2} (\mathcal{V}_{1})_{3\zeta} - 40\mathcal{V}_{0}\mathcal{V}_{3} (\mathcal{V}_{2})_{2\zeta} + 2\mathcal{V}_{0}\mathcal{V}_{3} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{0}\mathcal{V}_{0} (\mathcal{V}_{3})_{\zeta} + (\mathcal{V}_{2})_{2\zeta} (\mathcal{V}_{2})_{\zeta} \\ &\quad + 2\mathcal{V}_{1}\mathcal{V}_{2} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{0}\mathcal{V}_{3} (\mathcal{V}_{2})_{2\zeta} + 2\mathcal{V}_{0}\mathcal{V}_{3} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{0}\mathcal{V}_{4} (\mathcal{V}_{0})_{2\zeta} + (\mathcal{V}_{2})_{2\zeta} (\mathcal{V}_{2})_{\zeta} \\ &\quad + 2\mathcal{V}_{1}\mathcal{V}_{2} (\mathcal{V}_{1})_{2\zeta} + 2\mathcal{V}_{1}\mathcal{V}_{3} (\mathcal{V}_{0})_{2\zeta} + \mathcal{V}_{2}^{2} (\mathcal{V}_{0})_{\zeta} + (\mathcal{V}_{0})_{\zeta} (\mathcal{V}_{4})_{\zeta} - 40\mathcal{V}_{0}\mathcal{V}_{1} (\mathcal{V}_{3})_{3\zeta} - 40\mathcal{V}_{0}\mathcal{V}_{2} (\mathcal{V}_{2})_{3\zeta} - 40\mathcal{V}_{0}\mathcal{V}_{3} (\mathcal{V}_{1})_{3\zeta} \\ &\quad - 40\mathcal{V}_{0}\mathcal{V}_{4} (\mathcal{V}_{0})_{3\zeta} - 20\mathcal{V}_{1}^{2} (\mathcal{V}_{2})_{3\zeta} - 40\mathcal{V}_{1}\mathcal{V}_{2} (\mathcal{V}_{1})_{3\zeta} - 40\mathcal{V}_{0}\mathcal{V}_{2} (\mathcal{V}_{2})_{\zeta} \\ &\quad + (\mathcal{V}_{1})_{2\zeta} (\mathcal{V}_{3})_{\zeta} - 20\mathcal{V}_{1}^{2} (\mathcal{V}_{2})_{3\zeta} - 40\mathcal{V}_{1}\mathcal{V}_{2} (\mathcal{V}_{1})_{3\zeta} - 40\mathcal{V}_{0}\mathcal$$

Comparing the P-like coefficients, we have

$$\begin{split} P^{0}: \mathcal{V}_{0}\left(\zeta,\tau\right) &= \frac{1}{\zeta}, \\ P^{1}: \mathcal{V}_{1}\left(\zeta,\tau\right) &= -L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \Big[H_{0}\left(\mathcal{V}\right) + \left(\mathcal{V}_{0}\right)_{\zeta} + \left(\mathcal{V}_{0}\right)_{5\zeta} \Big] \Big] = \frac{\left(\tau^{\rho}/\rho\right)^{\beta}}{\zeta^{2} \Gamma\left(\beta+1\right)}, \\ P^{2}: \mathcal{V}_{2}\left(\zeta,\tau\right) &= -L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \Big[H_{1}\left(\mathcal{V}\right) + \left(\mathcal{V}_{1}\right)_{\zeta} + \left(\mathcal{V}_{1}\right)_{5\zeta} \Big] \bigg] = \frac{\left(\tau^{\rho}/\rho\right)^{2\beta}}{\zeta^{3} \Gamma\left(2\beta+1\right)} \\ P^{3}: \mathcal{V}_{3}\left(\zeta,\tau\right) &= -L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \Big[H_{2}\left(\mathcal{V}\right) + \left(\mathcal{V}_{2}\right)_{\zeta} + \left(\mathcal{V}_{2}\right)_{5\zeta} \Big] \bigg] = \frac{\left(\tau^{\rho}/\rho\right)^{3\beta}}{\zeta^{4} \Gamma\left(3\beta+1\right)} \\ P^{4}: \mathcal{V}_{4}\left(\zeta,\tau\right) &= -L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \Big[H_{3}\left(\mathcal{V}\right) + \left(\mathcal{V}_{3}\right)_{\zeta} + \left(\mathcal{V}_{3}\right)_{5\zeta} \Big] \bigg] = \frac{\left(\tau^{\rho}/\rho\right)^{4\beta}}{\zeta^{5} \Gamma\left(4\beta+1\right)} \\ P^{5}: \mathcal{V}_{5}\left(\zeta,\tau\right) &= -L_{\rho}^{-1} \bigg[\frac{1}{s^{\beta}} L_{\rho} \Big[H_{4}\left(\mathcal{V}\right) + \left(\mathcal{V}_{4}\right)_{\zeta} + \left(\mathcal{V}_{4}\right)_{5\zeta} \Big] \bigg] = \frac{\left(\tau^{\rho}/\rho\right)^{5\beta}}{\zeta^{6} \Gamma\left(5\beta+1\right)} \\ \vdots \end{split}$$

The analytical solution of $\mathcal{V}(\zeta, \tau)$ is defined as

$$\mathcal{V}(\zeta,\tau) = \sum_{i=0}^{\infty} \mathcal{V}(\zeta,\tau)_{i} = \frac{1}{\zeta} + \frac{(\tau^{\rho}/\rho)^{\beta}}{\zeta^{2}\Gamma(\beta+1)} + \frac{(\tau^{\rho}/\rho)^{2\beta}}{\zeta^{3}\Gamma(2\beta+1)} + \frac{(\tau^{\rho}/\rho)^{3\beta}}{\zeta^{4}\Gamma(3\beta+1)} + \frac{(\tau^{\rho}/\rho)^{4\beta}}{\zeta^{5}\Gamma(4\beta+1)} + \frac{(\tau^{\rho}/\rho)^{5\beta}}{\zeta^{6}\Gamma(5\beta+1)} + \cdots.$$
(27)

Then, put $\beta = 1$ in (27):

$$\mathscr{V}(\zeta,\tau) = \sum_{i=0}^{\infty} \mathscr{V}_i(\zeta,\tau) = \frac{1}{\zeta} + \frac{\tau}{\zeta^2} + \frac{\tau^2}{\zeta^3} + \frac{\tau^3}{\zeta^4} + \cdots$$
(28)

The exact result is $\mathscr{V}(\zeta, \tau) = 1/\zeta - \tau$.

In Figure 1, the three-dimensional figures of ρ -HPTM and exact results in graphs (a) and (b) respectively at $\beta = 1$ and the close contact of the exact and ρ -HPTM solutions are investigated. In Figure 2, represent that various fractional

(26)



FIGURE 1: Graph of (a) exact and (b) analytic solutions of $\beta = 1$ of Example 1.



FIGURE 2: Figure of various fractional orders of Example 1.

order of ρ -HPTM results at $\beta = 1, 0.8, 0.6, 0.4$. The nonclassical results are investigated to be converge to an integerorder result of the given problem.

Applying the ρ -Laplace transform on (29), we get

$$L_{\rho}[\mathcal{V}(\zeta,\tau)] = \frac{1}{s}e^{\zeta} + \frac{1}{s^{\beta}}L_{\rho}\left[\mathcal{VV}_{3\zeta} - \mathcal{VV}_{\zeta} - \mathcal{V}_{5\zeta}\right].$$
 (31)

Example 2. Consider the fifth-order nonlinear fraction KdV equation

$$D^{\beta}_{\tau}\mathcal{V} + \mathcal{V}\mathcal{V}_{\zeta} - \mathcal{V}\mathcal{V}_{3\zeta} + \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \le 1,$$
(29)

with the IC

$$\mathscr{V}(\zeta,\tau) = e^{\zeta}.$$
 (30)

$$L_{\rho}[\mathcal{V}(\zeta,\tau)] = \frac{1}{2}e^{\zeta} + \frac{1}{\beta}L_{\rho}[\mathcal{VV}_{3\zeta} - \mathcal{VV}_{\zeta} - \mathcal{V}_{5\zeta}].$$
(3)

Next, using the inverse of ρ -Laplace transform of (31),

$$\mathcal{V}(\zeta,\tau) = e^{\zeta} + L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ \mathcal{V} \mathcal{V}_{3\zeta} - \mathcal{V} \mathcal{V}_{\zeta} - \mathcal{V}_{5\zeta} \right\} \right].$$
(32)

Now, we apply HPM

$$\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) = e^{\zeta} + p \left[L_{\rho}^{-1} \left\{ \frac{1}{s^{\beta}} L_{\rho} \left(\left(\sum_{n=0}^{\infty} p^{n} H_{n}(\mathcal{V}) \right) - \left(\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) \right) \right)_{5\zeta} \right\} \right], \tag{33}$$

where $H_n(x)$ represents the nonlinear term of He's polynomial. For the first few components, we present He's polynomials

$$\begin{split} H_{0}(\mathcal{V}) &= \mathcal{V}_{o}(\mathcal{V}_{0})_{3\zeta} - \mathcal{V}_{0}(\mathcal{V}_{0})_{\zeta}, \\ H_{1}(\mathcal{V}) &= \mathcal{V}_{1}(\mathcal{V}_{0})_{3\zeta} + \mathcal{V}_{0}(\mathcal{V}_{1})_{3\zeta} - \mathcal{V}_{1}(\mathcal{V}_{0})_{\zeta} - \mathcal{V}_{0}(\mathcal{V}_{1})_{\zeta}, \\ H_{2}(\mathcal{V}) &= \mathcal{V}_{2}(\mathcal{V}_{0})_{3\zeta} + \mathcal{V}_{1}(\mathcal{V}_{1})_{3\zeta} + \mathcal{V}_{0}(\mathcal{V}_{2})_{3\zeta} \\ &- \mathcal{V}_{2}(\mathcal{V}_{0})_{\zeta} - \mathcal{V}_{1}(\mathcal{V}_{1})_{\zeta} - \mathcal{V}_{0}(\mathcal{V}_{2})_{\zeta}, \\ H_{3}(\mathcal{V}) &= \mathcal{V}_{3}(\mathcal{V}_{0})_{3\zeta} + \mathcal{V}_{2}(\mathcal{V}_{1})_{3\zeta} + \mathcal{V}_{1}(\mathcal{V}_{2})_{3\zeta} \\ &+ \mathcal{V}_{0}(\mathcal{V}_{3})_{3\zeta} - \mathcal{V}_{3}(\mathcal{V}_{0})_{\zeta} - \mathcal{V}_{2}(\mathcal{V}_{1})_{\zeta} \\ &- \mathcal{V}_{1}(\mathcal{V}_{2})_{\zeta} - \mathcal{V}_{0}(\mathcal{V}_{3})_{\zeta}, \\ H_{4}(\mathcal{V}) &= \mathcal{V}_{4}(\mathcal{V}_{0})_{3\zeta} + \mathcal{V}_{3}(\mathcal{V}_{1})_{3\zeta} + \mathcal{V}_{2}(\mathcal{V}_{2})_{3\zeta} \\ &+ \mathcal{V}_{1}(\mathcal{V}_{3})_{3\zeta} + \mathcal{V}_{0}(\mathcal{V}_{4})_{3\zeta} - \mathcal{V}_{4}(\mathcal{V}_{0})_{\zeta} \\ &- \mathcal{V}_{3}(\mathcal{V}_{1})_{\zeta} - \mathcal{V}_{2}(\mathcal{V}_{2})_{\zeta} - \mathcal{V}_{1}(\mathcal{V}_{3})_{\zeta} - \mathcal{V}_{0}(\mathcal{V}_{4})_{\zeta}, \\ \vdots \end{split}$$

$$(34)$$

Comparing the *P*-like coefficients, we have $p^0: \mathcal{V}_0(\zeta, \tau) = e^{\zeta}$,

$$p^{1}: \mathscr{V}_{1}(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ H_{0}(\mathscr{V}) - (\mathscr{V}_{0})_{5\zeta} \right\} \right] = -\frac{(\tau^{\rho}/\rho)^{\beta}}{\Gamma(\beta+1)} e^{\zeta},$$

$$p^{2}: \mathscr{V}_{2}(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ H_{1}(\mathscr{V}) - (\mathscr{V}_{1})_{5\zeta} \right\} \right] = \frac{(\tau^{\rho}/\rho)^{2\beta}}{\Gamma(2\beta+1)} e^{\zeta},$$

$$p^{3}: \mathscr{V}_{3}(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ H_{2}(\mathscr{V}) - (\mathscr{V}_{2})_{5\zeta} \right\} \right] = -\frac{(\tau^{\rho}/\rho)^{3\beta}}{\Gamma(3\beta+1)} e^{\zeta}, \quad (35)$$

$$p^{4}: \mathscr{V}_{4}(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ H_{3}(\mathscr{V}) - (\mathscr{V}_{3})_{5\zeta} \right\} \right] = \frac{(\tau^{\rho}/\rho)^{4\beta}}{\Gamma(4\beta+1)} e^{\zeta},$$

$$p^{5}: \mathscr{V}_{5}(\zeta, \tau) = L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left\{ H_{4}(\mathscr{V}) - (\mathscr{V}_{4})_{5\zeta} \right\} \right] = -\frac{(\tau^{\rho}/\rho)^{5\beta}}{\Gamma(5\beta+1)} e^{\zeta}.$$

$$\vdots$$

Therefore, the analytic solution of $\mathcal{V}(\zeta, \tau)$ is defined as

$$\mathcal{V}(\zeta,\tau) = \sum_{i=0}^{\infty} \mathcal{V}_i(\zeta,\tau) = e^{\zeta} \left(1 - \frac{(\tau^{\rho}/\rho)^{\beta}}{\Gamma(\beta+1)} + \frac{(\tau^{\rho}/\rho)^{2\beta}}{\Gamma(2\beta+1)} - \frac{(\tau^{\rho}/\rho)^{3\beta}}{\Gamma(3\beta+1)} + \frac{(\tau^{\rho}/\rho)^{4\beta}}{\Gamma(4\beta+1)} - \frac{(\tau^{\rho}/\rho)^{5\beta}}{\Gamma(5\beta+1)} + \cdots \right).$$
(36)

Then, $\beta = 1$ for (36), and we get

$$\mathscr{V}(\zeta,\tau) = \sum_{i=0}^{\infty} \mathscr{V}_i(\zeta,\tau) = e^{\zeta} \left(1 - \tau + \frac{\tau^2}{2!} - \frac{\tau^3}{3!} + \frac{\tau^4}{4!} - \frac{\tau^5}{5!} + \cdots \right).$$
(37)

The exact solution is $\mathscr{V}(\zeta, \tau) = e^{\zeta - \tau}$.

In Figure 3, the three-dimensional figures of ρ -HPTM and exact results in graphs (a) and (b) respectively at $\beta = 1$ and the close contact of the exact and ρ -HPTM solutions are investigated. In Figure 4, represent that various fractional order of ρ -HPTM results at $\beta = 1, 0.8, 0.6, 0.4$. The non-classical results are investigated to be converge to an integer-order result of the given problem.

Example 3. Consider nonlinear fractional-order Kawahara equation

$$D^{\beta}_{\tau}\mathcal{V} + \mathcal{V}\mathcal{V}_{\zeta} + \mathcal{V}_{3\zeta} - \mathcal{V}_{5\zeta} = 0, \quad 0 < \beta \le 1,$$
(38)

with the IC

$$\mathscr{V}(\zeta,\tau) = \frac{105}{169} \operatorname{sech}^4\left(\frac{\zeta-\phi}{2\sqrt{13}}\right). \tag{39}$$

Applying the ρ -Laplace transform on (38), we get

$$L_{\rho}\mathcal{V}(\zeta,\tau)] = \frac{1}{s} \frac{105}{169} \operatorname{sech}^{4} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) + \frac{1}{s^{\beta}} L_{\rho} \left[\mathcal{V}_{5\zeta} - \mathcal{V}_{3\zeta} - \mathcal{V}\mathcal{V}_{\zeta}\right].$$
(40)

Next, using the inverse of ρ -Laplace transform of (40),

$$\mathcal{V}(\zeta,\tau) = \frac{105}{169} \operatorname{sech}^4\left(\frac{\zeta-\phi}{2\sqrt{13}}\right) + L_{\rho}^{-1} \left[\frac{1}{s^{\beta}} L_{\rho} \left[\mathcal{V}_{5\zeta} - \mathcal{V}_{3\zeta} - \mathcal{V}\mathcal{V}_{\zeta}\right]\right].$$
(41)

Now, we apply HPM

$$\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) = \frac{105}{169} \operatorname{sech}^{4} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) + p \left[L_{\rho}^{-1} \left\{ \frac{1}{s^{\beta}} L_{\rho} \left(\left(\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) \right)_{5\zeta} - \left(\sum_{n=0}^{\infty} p^{n} \mathcal{V}_{n}(\zeta,\tau) \right)_{5\zeta} - \left(\sum_{n=0}^{\infty} p^{n} \mathcal{H}_{n}(\mathcal{V}) \right) \right) \right\} \right],$$

$$(42)$$

where $H_n(\mathcal{V})$ represent the nonlinear terms of He's polynomial. For the first few components, we present He's polynomials

$$\begin{split} H_0(\mathcal{V}) &= \mathcal{V}_0(\mathcal{V}_0)_{\zeta}, \\ H_1(\mathcal{V}) &= \mathcal{V}_0(\mathcal{V}_1)_{\zeta} + \mathcal{V}_1(\mathcal{V}_0)_{\zeta}, \\ H_2(\mathcal{V}) &= \mathcal{V}_0(\mathcal{V}_2)_{\zeta} + \mathcal{V}_1(\mathcal{V}_1)_{\zeta} + \mathcal{V}_2(\mathcal{V}_0)_{\zeta}, \\ H_3(\mathcal{V}) &= \mathcal{V}_0(\mathcal{V}_3)_{\zeta} + \mathcal{V}_1(\mathcal{V}_2)_{\zeta} + \mathcal{V}_2(\mathcal{V}_1)_{\zeta} + \mathcal{V}_3(\mathcal{V}_0)_{\zeta}, \\ H_4(\mathcal{V}) &= \mathcal{V}_0(\mathcal{V}_4)_{\zeta} + \mathcal{V}_1(\mathcal{V}_3)_{\zeta} + \mathcal{V}_2(\mathcal{V}_2)_{\zeta} \\ &\quad + \mathcal{V}_3(\mathcal{V}_1)_{\zeta} + \mathcal{V}_4(\mathcal{V}_0)_{\zeta}, \\ &\vdots \end{split}$$

(43)

Comparing the P-like coefficients, we get



FIGURE 3: Graph of (a) exact and (b) analytic solutions of $\beta = 1$ of Example 2.

$$\begin{split} p^{0} \colon \mathscr{V}_{0}(\zeta,\tau) &= \frac{105}{169} \operatorname{sech}^{4} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right), \\ p^{1} \colon \mathscr{V}_{1}(\zeta,\tau) &= L_{p}^{-1} \left[\frac{1}{s^{\beta}} L_{p} \left[(\mathscr{V}_{0})_{5\zeta} - (\mathscr{V}_{0})_{5\zeta} - H_{0}(\mathscr{V}) \right] \right] &= -\frac{100}{377\sqrt{13}} \operatorname{sech}^{4} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \operatorname{tanh} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \frac{(\tau^{p}/\rho)^{\beta}}{\Gamma(\beta + 1)}, \\ p^{2} \colon \mathscr{V}_{2}(\zeta,\tau) &= L_{p}^{-1} \left[\frac{1}{s^{\beta}} L_{p} \left[(\mathscr{V}_{1})_{5\zeta} - (\mathscr{V}_{1})_{3\zeta} - H_{1}(\mathscr{V}) \right] \right] \\ &= -\frac{21687}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{\delta} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \left[-3 + 2\operatorname{cosh} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \right] \frac{(\tau^{p}/\rho)^{2\beta}}{\Gamma(2\beta + 1)}, \\ p^{3} \colon \mathscr{V}_{3}(\zeta,\tau) &= L_{p}^{-1} \left[\frac{1}{s^{\beta}} L_{p} \left\{ (\mathscr{V}_{2})_{5\zeta} - (\mathscr{V}_{2})_{3\zeta} - H_{2}(\mathscr{V}) \right\} \right] \\ &= -\frac{461962}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{7} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \times \left[-13\operatorname{sinh} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) + 2\operatorname{sinh} \left(\frac{3(\zeta - t\phi)}{2\sqrt{13}} \right) \right] \frac{(\tau^{p}/\rho)^{3\beta}}{\Gamma(3\beta + 1)}, \\ p^{4} \colon \mathscr{V}_{4}(\zeta,\tau) &= L_{p}^{-1} \left[\frac{1}{s^{\beta}} L_{p} \left[(\mathscr{V}_{3})_{5\zeta} - (\mathscr{V}_{3})_{3\zeta} - H_{3}(\mathscr{V}) \right] \right] \\ &= -\frac{3784854}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{8} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \times \left[-49\operatorname{scosh} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) + 4\operatorname{cosh} \left(\frac{2(\zeta - t\phi)}{2\sqrt{13}} \right) + 52 \right] \frac{(\tau^{p}/\rho)^{4\beta}}{\Gamma(4\beta + 1)}, \\ p^{5} \colon \mathscr{V}_{5}(\zeta,\tau) &= L_{p}^{-1} \left[\frac{1}{s^{\beta}} L_{p} \left[(\mathscr{V}_{4})_{5\zeta} - (\mathscr{V}_{4})_{3\zeta} - H_{4}(\mathscr{V}) \right] \right] \\ &= -\frac{3.22496310 \times 10^{7}}{\sqrt{13}} \operatorname{sech}^{9} \left(\frac{\zeta - \phi}{2\sqrt{13}} \right) \times \left[171\operatorname{sinh} \left(\frac{3(\zeta - \phi)}{2\sqrt{13}} \right) - 8\operatorname{sinh} \left(\frac{5(\zeta - \phi)}{2\sqrt{13}} \right) \\ - \operatorname{661\operatorname{sinh}} \left(\frac{5(\zeta - t\phi)}{2sqrt13} \right) \right] \frac{(\tau^{p}/\rho)^{5\beta}}{\Gamma(5\beta + 1)}. \end{split}$$



FIGURE 4: Figure of (a) and (b) at various fractional-order of Example 2.

The analytic solution $\mathscr{V}(\zeta, \tau)$ is achieved as

$$\begin{aligned} \mathscr{V}(\zeta,\tau) &= \sum_{i=0}^{\infty} \mathscr{V}_{i}(\zeta,\tau), \\ \mathscr{V}(\zeta,\tau) &= \frac{105}{169} \operatorname{sech}^{4} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) - \frac{100}{377\sqrt{13}} \operatorname{sech}^{4} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \operatorname{tanh} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \frac{(\tau^{\rho}/\rho)^{\beta}}{\Gamma(\beta+1)} \\ &- \frac{21687}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{6} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \left[-3 + 2\operatorname{cosh} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \right] \frac{(\tau^{\rho}/\rho)^{2\beta}}{\Gamma(2\beta+1)} \\ &- \frac{461962}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{7} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \times \left[-13\operatorname{sinh} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) + 2\operatorname{sinh} \left(\frac{3(\zeta-\phi)}{2\sqrt{13}}\right) \right] \frac{(\tau^{\rho}/\rho)^{3\beta}}{\Gamma(3\beta+1)} \end{aligned} \tag{45} \\ &- \frac{3784854}{10 \times 10^{7}\sqrt{13}} \operatorname{sech}^{8} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \times \left[-49\operatorname{scosh} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) + 4\operatorname{cosh} \left(\frac{2(\zeta-\phi)}{2\sqrt{13}}\right) + 52 \right] \frac{(\tau^{\rho}/\rho)^{4\beta}}{\Gamma(4\beta+1)} \\ &- \frac{3.22496310 \times 10^{7}}{\sqrt{13}} \operatorname{sech}^{9} \left(\frac{\zeta-\phi}{2\sqrt{13}}\right) \times \left[171\operatorname{sinh} \left(\frac{3(\zeta-\phi)}{2\sqrt{13}}\right) - 8\operatorname{sinh} \left(\frac{5(\zeta-\phi)}{2\sqrt{13}}\right) \right] \\ &- 661\operatorname{sinh} \left(\frac{5(\zeta-\phi)}{2\sqrt{13}}\right) \right] \frac{(\tau^{\rho}/\rho)^{5\beta}}{\Gamma(5\beta+1)} + \cdots. \end{aligned}$$

The exact solution is $\mathcal{V}(\zeta, \tau) = 105/169 \text{sec}h^4 [1/2\sqrt{13} (\zeta + 36\tau/169 - \phi)].$

In Figure 5, the three-dimensional figures of ρ -HPTM and exact results in graphs (a) and (b) respectively at $\beta = 1$



FIGURE 5: Graph of (a) exact and (b) analytical results of $\beta = 1$ of Example 3.



FIGURE 6: Figure of (a) at various fractional-order of β and (b) error graph of Example 3.

and the close contact of the exact and ρ -HPTM solutions are investigated. In Figure 6, represent that various fractional order of ρ -HPTM results at $\beta = 1, 0.8, 0.6, 0.4$. The nonclassical results are investigated to be converge to an integer-order result of the given problem.

5. Conclusions

This paper determined the fractional-order Kawahara and fifth-order KdV equations, applying the ρ -homotopy perturbation transform method. The present method is

used to describe the results for specific examples. The ρ -HPTM result is highly congruent with the precise solution of the suggested problems. Additionally, the proposed method estimated the results of the cases using fractional-order derivatives. The graphical examination of the resulting fractional-order results proved their convergence to integer-order outcomes. Additionally, the ρ -HPTM technique is straightforward, simple, and computationally efficient; the suggested method can be adapted to solve additional fractional-order partial differential equations.

Data Availability

The numerical data used to support the findings of this study are included within the article.

Conflicts of Interest

The authors declare that there are no conflicts of interest regarding the publication of this article.

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Research Article

Persistence of Heteroclinic Cycles Connecting Repellers in Banach Spaces

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This paper is concerned with persistence of heteroclinic cycles connecting repellers in Banach spaces. It is proved that if a map with a regular and nondegenerate heteroclinic cycle connecting repellers undergoes a small perturbation, then the perturbed map can still have a regular and nondegenerate heteroclinic cycle connecting repellers. The perturbation rang is given by an explicit positive constant according to the properties of the original map. Hence, the perturbed map and the original map are simultaneously chaotic in the sense of both Devaney and Li-Yorke. Especially, the persistence of heteroclinic cycles connecting repellers is also discussed in the Euclidean space, where the repellers can expand in different norms. Finally, three examples are provided to illustrate the validity of the theoretical results.

1. Introduction

Chaos is a very important kind of dynamical behaviors in nonlinear systems and chaos problems have attracted a lot of attention from many scientists and sgd mathematicians. In 1975, the first mathematical definition of chaos and a famous result that "period three implies chaos" were given by Li and Yorke [1] in studying continuous interval maps. After that, different definitions of chaos from different points of view were proposed by researchers, one can see [2-4] for some related definitions of chaos. Among these mathematical definitions, chaos in the sense of Li-Yorke, Devaney or Wiggins are often used in the literature, see [5-8] for discussions of their relationships. Then, there appeared many works to study chaotic behaviors of multidimensional maps. A very famous work that "a snap-back repeller implies chaos" in the sense of Li-Yorke was proposed by Marotto [9], which is a generalization of Li and Yorke's result from onedimensional maps to multidimensional maps. This result shows great power in studying chaos of finite dimensional maps. However, it is clear that there are many systems should be studied in infinite dimensional spaces, such as Banach spaces and metric spaces. Then, a lot of works have been done on chaotic behaviors of infinite dimensional

maps. Some of these important results were given by Shi and her cooperators. In 2004, Shi and Chen [10] extended the concept of snap-back repeller to metric spaces and obtained several criteria of chaos. Later, Shi and her cooperators developed the coupled-expansion theory and used it to study chaos, see [11–14] and references therein.

Structural stability of chaotic maps is a very important and interesting question, and many results have been achieved. Marotto first studied perturbations of maps with snap-back repellers in [15, 16], and proved that if a scalar system with a snap-back repeller undergoes a small perturbation, then the perturbation system will have a transversal homoclinic point and thus has chaotic behaviors. Later, there appeared several results about multidimensional perturbations of chaotic systems, see [17-19]. In 2009, Li and Lyu [20] proved that if a map with a snap-back repeller in \mathbb{R}^n undergoes a mall C^1 perturbation, then the perturbed map still has a snap-back repeller and consequently is chaotic in the sense of Li-Yorke. However, all the above perturbations of chaotic systems were made in finite dimensional spaces. In 2011, Chen et al. [21] studied the persistence of snap-back repellers under small C^1 perturbations in Banach spaces. In 2012, Zhang et al. [22] used a different method to study the persistence of snap-back repellers under small Lipschitz perturbations in Banach spaces. Moreover, Zhang and Shi [23] studied the persistence of coupled-expansion for timevarying systems under small time-varying perturbations in Banach spaces, and showed the persistence of snap-back repellers.

In 2006, Lin and Chen [24] gave a result that heteroclinical repellers imply chaos in the sense of Li-Yorke in \mathbf{R}^{n} . In their definition of heteroclinical repellers, there were some conditions given by the Jacobian matrices of a map. However, a map in a metric space may not have derivatives in general. In 2008, based on their work, Li et al. [25] grasped the essential meanings of the definition of heteroclinical repellers to extend it to general metric spaces without needing the continuity or continuous differentiability. For more intuitive to reflect the relationships of the repellers, they redefined it as a heteroclinic cycle connecting repellers and obtained several criteria of chaos. Later, they studied chaos induced by heteroclinic cycles connecting repellers in general Banach spaces [26], and used these results to study existence of chaos or chaotification problems [27]. This shows that the heteroclinic cycle connecting repellers has significant effects on chaos studying. Hence, it is worth studying whether a heteroclinic cycle connecting repellers has the persistence under small perturbations as that for a snap-back repeller. Recently, in 2020, Chen and Wu [28] studied the persistence of heteroclinic repellers in \mathbf{R}^n for C^1 maps under small C^1 perturbations. Chen et al. [29] studied the persistence of heteroclinic repellers in Banach spaces for C^1 maps under small C^1 perturbations. It is noted that the definitions of heteroclinic repellers in [28, 29] both needed the differentiability of a map as that definition in [24]. In 2021, Wu [30] extended the concept of heteroclinic repellers in [24] to heteroclinic cycle connecting expanding periodic points in \mathbf{R}^n and studied the persistence of it for C^1 maps under C^1 perturbations, where the maps needed to be continuously differentiable in the whole space. More recently, Chen and Luo [31] studied the persistence of regular nondegenerate snap-back repellers and heteroclinic cycles for continuous maps under small Lipschitz perturbations, where the maps were continuous in the whole Banach space. On the one hand, it should be pointed out that all the above results needed the maps to be continuous or continuously differentiable in the whole space. However, there are a lot of maps that may not be continuous or continuously differentiable in the whole space. On the other hand, it should be pointed out that all the above results needed the perturbations to be small enough and did not give a relatively explicit expression for the range of small perturbations, which is convenient and useful in applications to quickly check out whether the persistence is maintained. So, it is meaningful to study persistence of heteroclinic cycles connecting repellers for maps which are only continuous or continuously differentiable in some domains of the whole space, and it is also meaningful to study the explicit expression for the range of small perturbations.

The fixed point theory has become an essential tool to resolve some problems in nonlinear analysis, including fractional calculus, see [32, 33] and references therein for more details about this theory. Here, we will apply the Banach contractive mapping principle and the ideas used in [22, 23] to study the persistence of regular and nondegenerate heteroclinic cycles connecting repellers in Banach spaces, where the original maps are only continuous or continuously differentiable in some neighborhoods of points. An important result is that an explicit expression for the range of perturbations is given. It will be proved that if a map with a regular and nondegenerate heteroclinic cycle connecting repellers undergoes a small Lipschitz perturbation, then the perturbed map can still have a regular and nondegenerate heteroclinic cycle connecting repellers. So, the perturbed map and the original map are simultaneously chaotic in the sense of both Devaney and Li-Yorke. Particularly, the persistence of heteroclinic cycles connecting repellers is also discussed in \mathbb{R}^n . The significant difference between our result and those obtained in [28, 30] is that the repellers in our result expand in different norms, while the repellers in the latter expand in the single Euclidean norm. It is clear that different fixed points can expand in different norms in \mathbb{R}^n . So, our result is more general in practice.

The rest of the paper is organized as follows. Some concepts and lemmas are given in Section 2. Several theorems about perturbations of maps with heteroclinic cycles connecting repellers in general Banach spaces or the Euclidean space are given in Section 3. Three examples are provided to illustrate the validity the theoretical results in Section 4. Finally, conclusions are made in Section 5.

2. Preliminaries

Some definitions and lemmas are given in this section.

Two usually used definitions of chaos in the sense of Li-Yoke or Devaney are first introduced. Then, the concept of a heteroclinic cycle connecting repellers is introduced.

Definition 1 (see [1]). Let (X, d) be a metric space, $f: X \longrightarrow X$ be a map, and S be a set of X with at least two distinct points. Then, S is called a scrambled set of f if for any two distinct points $x, y \in S$,

$$\liminf_{n \to \infty} d\left(f^n(x), f^n(y)\right) = 0, \quad \limsup_{n \to \infty} d\left(f^n(x), f^n(y)\right) > 0.$$
(1)

The map f is said to be chaotic in the sense of Li-Yorke if there exists an uncountable scrambled set S of f.

Remark 1. There are three conditions in the original characterization of chaos in Li-Yorke's theorem [1]. Since the third one is not essential, it is removed in Definition 1 in most literature.

Example 1. Consider the following Baker's equation

$$x_{n+1} = \begin{cases} 2x_n, & \text{for } 0 \le x_n \le \frac{1}{2}, \\ \\ 2(1-x_n), & \text{for } \frac{1}{2} < x_n \le 1, \end{cases}$$
(2)

which models the mixing of a dye spot on a strip of dough that is repeatedly stretched and folded over on itself. The iterative scheme (2) maps the interval [0, 1] into itself. It is easy to check that system (2) has a cycle of period three and hence is chaotic in the sense of Li-Yorke by the Li-Yorke theorem in [1]. This equation has been extensively discussed in the literature [7, 34] and references cited therein.

Definition 2 (see [4]). Let (X, d) be a metric space. A map $f: V \subset X \longrightarrow V$ is said to be chaotic on V in the sense of Devaney if

- (i) The set of the periodic points of f is dense in V
- (ii) f is topologically transitive in V
- (iii) f has sensitive dependence on initial conditions in V

Remark 2. In 1992, Banks et al. [5] proved that conditions (i) and (ii) together imply condition (iii) if f is continuous in V. So, condition (iii) is redundant in the above definition in this case. It has been proved by [6] that chaos in the sense of Devaney is stronger than chaos in the sense of Li-Yorke under some conditions.

Example 2. Let

$$\sum_{2}^{+} := \{ s = (s_0, s_1, s_2, \ldots) : s_j = 0 \text{ or } 1 \},$$
(3)

with the distance

$$\rho(s,t) := \sum_{i=0}^{\infty} \frac{|s_i - t_i|}{2^i},$$
(4)

where $s = (s_0, s_1, s_2, ...)$ and $t = (t_0, t_1, t_2, ...)$. Then $(\sum_{2}^{+} \rho)$ is a complete metric space and a Cantor set, see Lemma 2.5 in [10]. The shift map $\sigma: \Sigma_2^{+} \longrightarrow \Sigma_2^{+}$ defined by $\sigma((s_0, s_1, s_2, ...)) = (s_1, s_2, ...)$ is continuous. The dynamical system defined by σ is called a one-sided symbolic dynamical system. It follows from [[4], Part 1, Proposition 6.6] that σ has the following properties:

- (i) Card $\operatorname{Per}_n(\sigma) = 2^n$
- (ii) $Per(\sigma)$ is dense in Σ_2^+
- (iii) there exists a dense orbit of σ in Σ_2^+

Here, Card $\operatorname{Per}_n(\sigma)$ denotes the number of periodic points of period *n* for σ . It is clear that property (iii) implies that σ is transitive. Therefore, the symbolic dynamical system is chaotic in the sense of Devaney. See [3, 4] for more discussions about this symbolic dynamical system.

Definition 3 (see [26], Definition 2.5). Let (X, d) be a metric space and $f: X \longrightarrow X$ be a map with $k (\ge 2)$ fixed points $z_1, \ldots, z_k \in X$.

(I) Suppose that, for each $i (1 \le i \le k)$, z_i is an expanding fixed point of f in $\overline{B}_{r_i}(z_i)$, and there exist a point $x_{i0} \in B_{r_i}(z_i)$, $x_{i0} \ne z_i$, and a positive integer $m_i \ge 1$ such that $f^{m_i}(x_{i0}) = z_{t(i)}$, and z_i is the limit for the backward orbit of x_{i0} , where $\overline{B}_{r_i}(z_i)$ and $\overline{B}_{r_i}(z_i)$ are the closed and open balls of radius r_i centered at z_i , $t(i) = [i \mod k] + 1$. Then all the points

 x_{i0} ($1 \le i \le k$), together with their backward and forward orbits consist of a set, which is called a *k*-heteroclinic cycle connecting repellers z_1, \ldots, z_k .

(II) Suppose that f has a k-heteroclinic cycle connecting repellers z_1, \ldots, z_k . For each point x_0 on the cycle, if there exists a positive constant r_0 such that for each positive constant $r \le r_0$, $f(x_0)$ is an interior point of $f(B_r(x_0))$, then the cycle is called regular; if there exist positive constants r_1 and μ such that $d(f(x), f(y)) \ge \mu d(x, y), \forall x, y \in \overline{B}_{r_1}(x_0)$, then the cycle is called nondegenerate.

Remark 3. It is pointed out that the necessary and sufficient condition for a heteroclinic cycle connecting repellers is used to give the definition (I) for simplicity, see (1) of Remark 2.2 in [26]. In addition, it does not need the continuity or continuous differentiability in this definition, while some similar definitions need them, see [24–26, 28–31] for more details about this concept.

For convenience, some notations are given in the following. The continuously differentiable maps in a set U of a Banach space X are denoted by $C^1(U, X)$. The derivative of a map f at a point $x \in X$ is denoted by Df(x). In addition, for a linear map $L: X \longrightarrow X$, denote

$$\|L\| := \sup\{\|Lx\|: x \in X, \|x\| = 1\},$$

$$\|L\|^{0} := \inf\{\|Lx\|: x \in X, \|x\| = 1\}.$$
(5)

If a bounded linear map L has a bounded inverse, then L is said to be an invertible linear map, see Definition 4.17 in [35]. The following four lemmas will be used in the paper.

Lemma 1 (see [22], Lemma 2.4). Let $(X, \|\cdot\|)$ be a Banach space, $z \in X$, and $f: \overline{B}_r(z) \longrightarrow f(\overline{B}_r(z))$ be a continuous map. Assume that $f(B_r(z))$ is an open set of X and

$$\|f(x) - f(y)\| \ge \mu \|x - y\|, \quad \forall x, y \in \overline{B}_r(z), \tag{6}$$

for some constant $\mu > 0$, then

$$B_{(\mu-L)r}(F(z)) \in F(B_r(z)), \tag{7}$$

where F = f + g and g is a Lipschitz map in $\overline{B}_r(z)$ with Lipschitz constant $L < \mu$.

Lemma 2 (see [25], Theorem 3.4). Let (X, d) be a complete metric space and $f: X \longrightarrow X$ be a map. Assume that

- (i) f has a regular and nondegenerate k-heteroclinic cycle connecting repellers $z_1, \ldots, z_k \in X, k \ge 2$
- (ii) f is continuous in some neighborhood of each point on the cycle

Then there exists an uncountable, perfect, bounded, and closed set V such that f(V) = V and f is chaotic on V in the sense of Devaney as well as in the sense of Li-Yorke.

Lemma 3 (see [26], Lemma 2.2; [22], Lemma 2.3). Let $(X, \|\cdot\|)$ be a Banach space and $f: X \longrightarrow X$ be a map. Assume that f has a heteroclinic cycle connecting repellers

 $z_1, \ldots, z_k \in X, k \ge 2$, and for each point x_0 on the cycle f is continuously differentiable in some neighborhood of x_0 and satisfies that $Df(x_0)$ is an invertible linear map, then the cycle is regular and nondegenerate.

Lemma 4 (see [11], Lemma 2.2). Let $(X, \|\cdot\|)$ be a Banach space. Suppose that a map $f: X \longrightarrow X$ is continuously differentiable in $B_{r_0}(x_0)$ for some $x_0 \in X$ and some $r_0 > 0$, and satisfies that $\lambda_0 = \|Df(x_0)\|^0 > 0$, then for each $\varepsilon \in (0, \lambda_0)$, there exists a positive constant $r_1 < r_0$ such that

$$\|f(x) - f(y)\| \ge (\lambda_0 - \varepsilon) \|x - y\|, \quad \forall x, y \in \overline{B}_{r_1}(x_0).$$
(8)

3. Persistence of Heteroclinic Cycles Connecting Repellers in Banach Spaces

In this section, we will study persistence of heteroclinic cycles connecting repellers in Banach spaces. Assume that $(X, \|\cdot\|)$ is a Banach space, $f, g: X \longrightarrow X$ are two maps, and f has a regular and nondegenerate heteroclinic cycle connecting repellers and is continuous in some neighborhoods of interest points. Here, we study the following system:

$$x_{n+1} = f(x_n) + g(x_n), \quad n \ge 0,$$
 (9)

where g is viewed as a mall perturbation. It is proved that there still has a regular and nondegenerate heteroclinic cycle connecting repellers in (9) when g satisfies some conditions. Consequently, the perturbed system (9) is chaotic in the sense of both Devaney and Li-Yorke.

Theorem 1. Suppose that $(X, \|\cdot\|)$ is a Banach space and $f: X \longrightarrow X$ is a map with $k (\geq 2)$ different fixed points $z_1, \ldots, z_k \in X$ and satisfies the following:

- (i) For each i $(1 \le i \le k)$, z_i is a regular expanding fixed point of f in $\overline{B}_{r_i}(z_i)$ with expanding coefficient λ_{i0} for some constant $r_i > 0$. Furthermore, there exist a point $x_{i0} \in B_{r_i}(z_i)$, $x_{i0} \ne z_i$, and a positive integer $m_i \ge 1$ such that $f^{m_i}(x_{i0}) = z_{t(i)}$, where $t(i) = [i \mod k] + 1$. Consequently, f has a heteroclinic cycle Γ connecting repellers z_1, \ldots, z_k .
- (ii) The heteroclinic cycle Γ connecting repellers is regular and nondegenerate, and f is continuous in $\overline{B}_{r_i}(z_i)$ and some neighborhood U_{ij} of x_{ij} , where $x_{ij} = f^j(x_{i0})$ for $1 \le i \le k$, $1 \le j \le m_i - 1$.

Then, there exists a constant $\varepsilon_0 > 0$ such that for any Lipschitz map g in each set of $\overline{B}_{r_i}(z_i)$ and U_{ij} , $1 \le i \le k$, $1 \le j \le m_i - 1$, with the Lipschitz constant L satisfying

$$\max\left\{L, \|g(z_i)\|, \|g(x_{ij})\|, 1 \le i \le k, 0 \le j \le m_i - 1\right\} < \varepsilon_0.$$
(10)

The perturbed system (9) also has a regular and nondegenerate heteroclinic cycle Γ' connecting repellers, and consequently there exists an uncountable, perfect, bounded, and closed set *V* such that system (9) is chaotic on *V* in the sense of both Devaney and Li-Yorke. *Proof.* Without loss of generality and for simplicity, we only show that Theorem 1 is true for k = 2. When k > 2, one can use a similar method to prove it. For convenience, let F(x) : = f(x) + g(x) in the rest of this paper and i = 1 or 2 in the rest of this proof.

Without loss of generality, we can suppose that $\overline{B}_{r_1}(z_1) \cap \overline{B}_{r_2}(z_2) = \emptyset$, and $f(x_{i0}) \notin B_{r_i}(z_i)$. Otherwise, one can see the third paragraph in the proof of Theorem 3.1 in [25].

Since z_i is a regular expanding fixed point of f in $\overline{B}_{r_i}(z_i)$ with an expanding coefficient λ_{i0} , we get that

$$\|f(x) - f(y)\| \ge \lambda_{i0} \|x - y\|, \quad \forall x, y \in \overline{B}_{r_i}(z_i), \tag{11}$$

 $f: \overline{B}_{r_i}(z_i) \longrightarrow f(\overline{B}_{r_i}(z_i))$ is a homeomorphism and $f(B_{r_i}(z_i))$ is open, f(D) is open for any open set $D \in B_{r_i}(z_i)$. Take a constant

$$\delta_{i0} < \frac{r_i - \|z_i - x_{i0}\|}{2},\tag{12}$$

such that $\overline{B}_{\delta_{i_0}}(x_{i_0}) \subset B_{r_i}(z_i)$. Then, it follows form (11) that $f: B_{\delta_{i_0}}(x_{i_0}) \longrightarrow f(B_{\delta_{i_0}}(x_{i_0}))$ is also a homeomorphism.

From assumption (ii), it follows that there exist positive constants μ_{ij} and δ_{ij} such that

$$\|f(x) - f(y)\| \ge \mu_{ij} \|x - y\|, \quad \forall x, y \in \overline{B}_{\delta_{ij}}(x_{ij}),$$
(13)

 $f: B_{\delta_{ij}}(x_{ij}) \longrightarrow f(B_{\delta_{ij}}(x_{ij}))$ is homeomorphic, and $f(B_{\delta_{ij}}(x_{ij}))$ is open for $1 \le j \le m_i - 1$, where δ_{ij} satisfies the following conditions

$$\delta_{i1} < \lambda_{i0} \delta_{i0}, \quad \delta_{i,j+1} < \mu_{ij} \delta_{ij}, \quad \text{for } 1 \le j \le m_i - 2, \tag{14}$$

 $\overline{B}_{\delta_{ij}}(x_{ij})$ are disjoint subsets of U_{ij} and $\overline{B}_{\delta_{ij}}(x_{ij}) \cap B_{r_i}(z_i) = \emptyset$ for fixed *i* and $1 \le j \le m_i - 1$.

In the following, we will show that the map F satisfies the conditions in Lemma 2. It will be finished by the following three steps.

Step 1. It is to prove that F has two regular expanding fixed points z_1^* and z_2^* when g satisfies some conditions.

For proving the existence of z_1^* , we take two positive constants δ_{2,m_2} and ε_1 such that

$$\delta_{2,m_{2}} < \begin{cases} \min\left\{\lambda_{20}\delta_{20}, \frac{r_{1} - \|z_{1} - x_{10}\|}{2} - \delta_{10}\right\}, & \text{if } m_{2} = 1; \\\\ \min\left\{\mu_{2,m_{2}-1}\delta_{2,m_{2}-1}, \frac{r_{1} - \|z_{1} - x_{10}\|}{2} - \delta_{10}\right\}, & \text{if } m_{2} > 1. \end{cases}$$

$$\varepsilon_{1} = \frac{(\lambda_{10} - 1)\delta_{2,m_{2}}}{1 + \delta_{2,m_{2}}}. \tag{15}$$

Consider the following equation

$$F(x) = x, \quad x \in \overline{B}_{\delta_{2,m_2}}(z_1), \tag{16}$$

which is equivalent to the following equation:

$$f(x) = x - g(x), \quad x \in \overline{B}_{\delta_{2,m_2}}(z_1).$$
(17)

It follows from the first relation of (15) that $\overline{B}_{\delta_{2,\underline{m}_{2}}}(z_{1}) \subset B_{r_{1}}(z_{1})$. By assumption (i) and (11), we get that $f: \overline{B}_{\delta_{2,m_{2}}}(z_{1}) \longrightarrow f(\overline{B}_{\delta_{2,m_{2}}}(z_{1}))$ is homeomorphic. Then we obtain that $f(B_{\delta_{2,m_{2}}}(z_{1}))$ is an open set and the inverse map $f^{-1}: f(\overline{B}_{\delta_{2,m_{2}}}(z_{1})) \longrightarrow \overline{B}_{\delta_{2,m_{2}}}(z_{1})$ satisfies the following: $\|f^{-1}(x) - f^{-1}(y)\| \leq \lambda_{10}^{-1} \|x - y\|, \quad \forall x, y \in f(\overline{B}_{\delta_{2,m_{2}}}(z_{1}))$. (18)

Hence, equation (17) is translated into the following:

$$f^{-1}(x - g(x)) = x.$$
(19)

Here, it should prove that

$$x - g(x) \in f\left(\overline{B}_{\delta_{2,m_2}}(z_1)\right), \quad \forall x \in \overline{B}_{\delta_{2,m_2}}(z_1).$$
 (20)

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On the one hand, for any
$$x \in \overline{B}_{\delta_{2,m_2}}(z_1)$$
, we have
 $g(x) \| = \|g(x) - g(z_1) + g(z_1)\| \le \|g(x) - g(z_1)\| + \|g(z_1)\| \le L\delta_{2,m_2} + \|g(z_1)\|.$
(21)

Suppose that the map g satisfies

$$\max\{L, \|g(z_1)\|\} < \varepsilon_1, \tag{22}$$

then,

$$L\delta_{2,m_2} + \|g(z_1)\| < \varepsilon_1 \delta_{2,m_2} + \varepsilon_1 = (\lambda_{10} - 1)\delta_{2,m_2}.$$
 (23)

Therefore, it follows from (21) and (23) that, for any $x \in \overline{B}_{\delta_{2,m_2}}(z_1)$,

$$\|x - g(x) - z_1\| \le \|g(x)\| + \|x - z_1\| \le (\lambda_{10} - 1)\delta_{2,m_2} + \delta_{2,m_2} = \lambda_{10}\delta_{2,m_2}.$$
(24)

On the other hand, for any
$$x \in \partial B_{\delta_{2,m_2}}(z_1)$$
,
 $\|f(x) - z_1\| = \|f(x) - f(z_1)\| \ge \lambda_{10} \|x - z_1\| = \lambda_{10} \delta_{2,m_2}$.
(25)

Since $z_1 \in f(B_{\delta_{2,m_2}}(z_1))$ and $f(B_{\delta_{2,m_2}}(z_1))$ is an open set, it follows from (24) and (25) that (20)² is true.

According to the above discussion, we can define a map

$$h_1(x) = f^{-1}(x - g(x)), \quad x \in \overline{B}_{\delta_{2,m_2}}(z_1).$$
 (26)

For any $x \in \overline{B}_{\delta_{2m}}(z_1)$, it follows from (18) and (24) that

$$\|h_1(x) - z_1\| = \|f^{-1}(x - g(x)) - f^{-1}(z_1)\| \le \lambda_{10}^{-1} \|x - g(x) - z_1\| < \delta_{2, m_2},$$
(27)

which implies that h_1 maps $\overline{B}_{\delta_{2,m_2}}(z_1)$ into itself. Moreover, for any $x, y \in \overline{B}_{\delta_{2,m_2}}(z_1)$, we get from (18) that

$$\|h_{1}(x) - h_{1}(y)\| = \|f^{-1}(x - g(x)) - f^{-1}(y - g(y))\|$$

$$\leq \lambda_{10}^{-1}[\|g(x) - g(y)\| + \|x - y\|] \qquad (28)$$

$$\leq \lambda_{10}^{-1}(L + 1)\|x - y\|.$$

It follows from the second relation of (15) and (22) that

$$\lambda_{10} > L + 1, \tag{29}$$

which together with (28) yields that h_1 is contractive in $\overline{B}_{\delta_{2,m_2}}(z_1)$. It follows from the Banach contractive mapping principle and (27) that there exists a unique point $z_1^* \in B_{\delta_{2,m_2}}(z_1)$ satisfying $h_1(z_1^*) = z_1^*$. Consequently, $F(z_1^*) = z_1^*$, that is, z_1^* is a fixed point of F in $B_{\delta_{2,m_2}}(z_1)$.

It should prove that z_1^* is a regular expanding fixed point of F in some neighborhood of z_1^* . To do this, take

$$r_1^* = \frac{r_1 + \left\| z_1 - x_{10} \right\|}{2}.$$
 (30)

Then, it follows from $z_1^* \in B_{\delta_{2,m_2}}(z_1)$ and the first relation of (15) that

$$\begin{aligned} \|x_{10} - z_1^*\| &= \|x_{10} - z_1 + z_1 - z_1^*\| \\ &\leq \|x_{10} - z_1\| + \|z_1 - z_1^*\| \\ &\leq \|x_{10} - z_1\| + \delta_{2,m_2} \\ &< \|x_{10} - z_1\| + \frac{r_1 - \|z_1 - x_{10}\|}{2} \\ &- \delta_{10} = r_1^* - \delta_{10}, \end{aligned}$$
(31)

which implies that $\overline{B}_{\delta_{10}}(x_{10}) \in B_{r_1^*}(z_1^*)$. For any $x \in B_{r_1^*}(z_1^*)$,

$$\begin{aligned} \|x - z_1\| &= \|x - z_1^* + z_1^* - z_1\| \le \|x - z_1^*\| + \|z_1^* - z_1\| \\ &< r_1^* + \delta_{2,m_2} < \frac{r_1 + \|z_1 - x_{10}\|}{2} + \frac{r_1 - \|z_1 - x_{10}\|}{2} - \delta_{10} \\ &= r_1 - \delta_{10}, \end{aligned}$$
(32)

which implies that $\overline{B}_{r_1^*}(z_1^*) \in B_{r_1}(z_1)$. Consequently, $f(B_{r_1^*}(z_1^*))$ is an open set. For any $x, y \in \overline{B}_{r_1^*}(z_1^*)$,

$$\|F(x) - F(y)\| = \|f(x) + g(x) - f(y) - g(y)\|$$

$$\geq \|f(x) - f(y)\| - \|g(x) - g(y)\| \ge (\lambda_{10} - L)\|x - y\|.$$
(33)

Then, it follows from (29) and (33) that z_1^* is an expanding fixed point of F in $\overline{B}_{r_1^*}(z_1^*)$ with expanding coefficient $\lambda_{10} - L > 1$. Since $f(B_{r_1^*}(z_1^*))$ is an open set, it follows from Lemma 1 that

$$B_{(\lambda_{10}-L)r_1^*}(z_1^*) = B_{(\lambda_{10}-L)r_1^*}(F(z_1^*)) \subset F\left(B_{r_1^*}(z_1^*)\right).$$
(34)

which implies that z_1^* is an interior point of $F(B_{r_1^*}(z_1^*))$. Hence, z_1^* is a regular fixed point of F in $\overline{B}_{r_1^*}(z_1^*)$.

Here, it is to show that $F(B_{r_1^*}(z_1^*))$ is an open set. For each given point $y \in F(B_{r_1^*}(z_1^*))$, there is a point $x \in B_{r_1^*}(z_1^*)$ satisfying F(x) = y. Then, there is a constant $\overline{r_1} > 0$ satisfying $B_{\overline{r_1}}(x) \subset B_{r_1^*}(z_1^*)$. From the third paragraph of the proof, it is easy to see that $f(B_{\overline{r_1}}(x))$ is an open set because of $B_{r_1^*}(z_1^*) \subset B_{r_1}(z_1)$. It also follows from Lemma 1 again that

$$B_{(\lambda_{10}-L)\bar{r}_{1}}(y) = B_{(\lambda_{10}-L)\bar{r}_{1}}(F(x)) \subset F(B_{\bar{r}_{1}}(x)) \subset F(B_{r_{1}^{*}}(z_{1}^{*})), \quad (35)$$

which implies that y is an interior point of $F(B_{r_1^*}(z_1^*))$ and then $F(B_{r_1^*}(z_1^*))$ is an open set.

With a similar argument to the existence of z_1^* , we can obtain the following positive constants

$$\delta_{1,m_{1}} < \begin{cases} \min \begin{cases} \lambda_{10} \delta_{10}, \\ \frac{r_{2} - ||z_{2} - x_{20}||}{2} - \delta_{20} \end{cases}, & \text{if } m_{1} = 1; \\ \min \begin{cases} \frac{\mu_{1,m_{1}-1} \delta_{1,m_{1}-1}, \\ \frac{r_{2} - ||z_{2} - x_{20}||}{2} - \delta_{20} \end{cases}, & \text{if } m_{1} > 1. \quad (36) \\ \varepsilon_{2} = \frac{(\lambda_{20} - 1) \delta_{1,m_{1}}}{1 + \delta_{1,m_{1}}}, \\ r_{2}^{*} = \frac{r_{2} + ||z_{2} - x_{20}||}{2}, \end{cases}$$

such that when

$$\max\{L, \|g(z_2)\|\} < \varepsilon_2, \tag{37}$$

there exists a point $z_2^* \in B_{\delta_{1,m_1}}(z_2)$ satisfying that z_2^* is a regular expanding fixed point of F in $(\overline{B}_{r_2^*}(z_2^*))$ with expanding coefficient $\lambda_{20} - L > 1$ and $F(B_{r_2^*}(z_2^*))$ is an open set.

A summary for this step is given as follows. When the following condition holds

$$\max\{L, \|g(z_1)\|, \|g(z_2)\|\} < \min\{\varepsilon_1, \varepsilon_2\},$$
(38)

the map *F* will have two regular expanding fixed points $z_1^* \in B_{r_1^*}(z_1^*)$ and $z_2^* \in B_{\delta_{1,m_1}}(z_2)$, and $F(B_{r_i^*}(z_i^*))$ is an open set for i = 1, 2.

Step 2. It is to show that for each $i(1 \le i \le 2)$ there exists a point y_{i0} in $B_{r_i^*}(z_i^*)$ such that $F^{m_i}(y_{i0}) = z_{t(i)}^*$, where $t(i) = [i \mod 2] + 1$.

We first prove that there exist m_1 points $y_{1j} \in B_{\delta_{1j}}(x_{1j})$, $0 \le j \le m_1 - 1$ such that

$$F(y_{1j}) = y_{1,j+1}, \quad \text{for } 0 \le j \le m_1 - 2,$$

$$F(y_{1,m_1-1}) = z_2^*.$$
(39)

That is, there exists a point $y_{10} \in B_{\delta_{10}}(x_{10}) \subset B_{r_1^*}(z_1^*)$ such that $F^{m_1}(y_{10}) = z_2^*$.

In order to do that, we first prove the existence of y_{1,m_1-1} by solving the following equation:

$$F(x) = z_2^*, \quad x \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1}), \tag{40}$$

which can be translated into the following:

$$f(x) = z_2^* - g(x), \quad x \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1}).$$
 (41)

It follows from assumption (ii) and (13) that $f: \overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1}) \longrightarrow f(\overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1}))$ is homeomorphic with the inverse map $f^{-1}: f(\overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1})) \longrightarrow \overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1})$ satisfying

$$\left\|f^{-1}(x) - f^{-1}(y)\right\| \le \mu_{1,m_{1}-1}^{-1} \|x - y\|, \quad \forall x, y \in f\left(\overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1})\right).$$

$$(42)$$

Then, equation (41) can be translated into the following:

$$f^{-1}(z_2^* - g(x)) = x, \quad x \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1}).$$
(43)

Here, it needs to prove that

$$z_{2}^{*} - g(x) \in f\left(\overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1})\right), \quad x \in \overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1}).$$

$$(44)$$

Suppose that g also satisfies

$$\max\left\{L, \left\|g\left(x_{1j}\right)\right\|, 0 \le j \le m_1 - 1\right\} < \varepsilon_3, \tag{45}$$

$$\varepsilon_{3} = \min\left\{\varepsilon_{1}, \varepsilon_{2}, \frac{\min\{\lambda_{10}\delta_{10} - \delta_{11}, \mu_{1j}\delta_{1j} - \delta_{1,j+1}, \text{ for } 1 \le j \le m_{1} - 1\}}{1 + \max\{\delta_{1j}, \ 0 \le j \le m_{1} - 1\}}\right\}.$$
(46)

where

From (14) and (36), we get that $\varepsilon_3 > 0$. On the one hand, for any $x \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1})$, it follows from (45) and (46) that

$$\begin{aligned} \|z_{2}^{*} - g(x) - z_{2}\| &= \left\|g(x_{1,m_{1}-1}) - g(x) - g(x_{1,m_{1}-1}) + z_{2}^{*} - z_{2}\right\| \\ &\leq L \|x - x_{1,m_{1}-1}\| + \left\|g(x_{1,m_{1}-1})\right\| + \|z_{2}^{*} - z_{2}\| \\ &< \varepsilon_{3}\delta_{1,m_{1}-1} + \varepsilon_{3} + \delta_{1,m_{1}} = (1 + \delta_{1,m_{1}-1})\varepsilon_{3} + \delta_{1,m_{1}} \\ &\leq (1 + \delta_{1,m_{1}-1})\frac{\mu_{1,m_{1}-1}\delta_{1,m_{1}-1} - \delta_{1,m_{1}}}{1 + \delta_{1,m_{1}-1}} + \delta_{1,m_{1}} = \mu_{1,m_{1}-1}\delta_{1,m_{1}-1}. \end{aligned}$$

$$(47)$$

On the other hand, for any $x \in \partial B_{\delta_{1,m_1-1}}(x_{1,m_1-1})$, it follows from (13) that

$$\left\| f(x) - z_2 \right\| = \left\| f(x) - f\left(x_{1,m_1-1}\right) \right\| \ge \mu_{1,m_1-1} \left\| x - x_{1,m_1-1} \right\| = \mu_{1,m_1-1} \delta_{1,m_1-1}.$$
(48)

Since $z_2 \in f(B_{\delta_{1,m_1-1}}(x_{1,m_1-1}))$ and $f(B_{\delta_{1,m_1-1}}(x_{1,m_1-1}))$ is open, it follows from (47) and (48) that (44) is true. So, we can define a map

$$h_{2}(x) = f^{-1}(z_{2}^{*} - g(x)), \quad x \in \overline{B}_{\delta_{1,m_{1}-1}}(x_{1,m_{1}-1}).$$
(49)

It follows from (45) and (46) that

$$L < \varepsilon_3 < \frac{\mu_{1,m_1-1}\delta_{1,m_1-1} - \delta_{1,m_1}}{1 + \delta_{1,m_1-1}} < \mu_{1,m_1-1} \frac{\delta_{1,m_1-1}}{1 + \delta_{1,m_1-1}} < \mu_{1,m_1-1}.$$
(50)

Then, for any $x \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1})$, it follows from (42) and (47) that

$$\begin{split} \left\| h_{2}(x) - x_{1,m_{1}-1} \right\| &= \left\| f^{-1} \left(z_{2}^{*} - g(x) \right) - f^{-1} \left(z_{2} \right) \right\| \\ &\leq \mu_{1,m_{1}-1}^{-1} \left\| z_{2}^{*} - g(x) - z_{2} \right\| < \delta_{1,m_{1}-1}. \end{split}$$
(51)

That is, h_2 maps $\overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1})$ into itself. Moreover, for any $x, y \in \overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1})$, it follows from (42) that

 $\|h_{2}(x) - h_{2}(y)\| = \|f^{-1}(z_{2}^{*} - g(x)) - f^{-1}(z_{2}^{*} - g(y))\|$ (52) $\leq \mu_{1,m,-1}^{-1} \|g(x) - g(y)\| \leq L \mu_{1,m,-1}^{-1} \|x - y\|,$

which together with (50) implies that h_2 is contractive in $\overline{B}_{\delta_{1,m_1-1}}(x_{1,m_1-1})$. It follows from the Banach contractive mapping principle and (51) that there exists a unique point $y_{1,m_1-1} \in B_{\delta_{1,m_1-1}}(x_{1,m_1-1})$ such that $h(y_{1,m_1-1}) = z_2^*$. Consequently, $F(y_{1,m_1-1}) = z_2^*$, that is, equation (40) has a unique solution $y_{1,m_1-1} \in B_{\delta_{1,m_1-1}}(x_{1,m_1-1})$. Using a similar method as above, we can prove that there

exist $m_1 - 1$ unique points $y_{1j} \in B_{\delta_{1j}}(x_{1j})$ for $0 \le j \le m_1 - 2$ such that $F(y_{1j}) = y_{1,j+1}$. Then, we get that $y_{10} \in B_{\delta_{10}}(x_{10}) \subset B_{r_1^*}(z_1^*)$ such that $F^{m_1}(y_{10}) = z_2^*$. Next, set g also to satisfy

$$\max\left\{L, \left\|g\left(x_{2j}\right)\right\|, 0 \le j \le m_2 - 1\right\} < \varepsilon_4, \tag{53}$$

where

$$\varepsilon_{4} = \min\left\{\varepsilon_{1}, \varepsilon_{2}, \frac{\min\left\{\lambda_{20}\delta_{20} - \delta_{21}, \mu_{2j}\delta_{2j} - \delta_{2,j+1}, \text{ for } 1 \le j \le m_{2} - 1\right\}}{1 + \max\left\{\delta_{2j}, \ 0 \le j \le m_{2} - 1\right\}}\right\}.$$
(54)

It follows from (14) and (15) that $\varepsilon_4 > 0$.

Repeating a similar discussion as above, we can get when g satisfies (53), there exist m_2 unique points $y_{2j} \in B_{\delta_{2j}}(x_{2j})$, $0 \le j \le m_2 - 1$, such that $F(y_{2j}) = y_{2,j+1}$ for $0 \le j \le m_2 - 2$, and $F(y_{2,m_2-1}) = z_1^*$. That is, there exists a point $y_{20} \in B_{\delta_{20}}(x_{20}) \subset B_{r_2^*}(z_2^*)$ such that $F^{m_2}(y_{20}) = z_1^*$.

Let $\tilde{\varepsilon}_0 = \min \{ \varepsilon_j, 1 \le j \le 4 \}$. If *g* satisfies the following condition

$$\max\left\{L, \left\|g(z_{i})\right\|, \left\|g(x_{ij})\right\|, 1 \le i \le 2, 0 \le j \le m_{i} - 1\right\} < \varepsilon_{0},$$
(55)

then the statements in Step 2 hold. Consequently, *F* has a heteroclinic cycle Γ' connecting repellers z_1^* and z_2^* .

Step 3. It is to show that the heteroclinic cycle Γ' connecting repellers z_1^* and z_2^* of *F* is regular and nondegenerate.

When the map g satisfies (55), it follows from the discussions in Step 2 that $y_{ij} \in B_{\delta_{ij}}(x_{ij})$ for $1 \le i \le 2$, $0 \le j \le m_i - 1$. Hence, for $1 \le i \le 2$, $0 \le j \le m_i - 1$, we can take positive constants $\delta'_{ij} < \delta_{ij}$ such that

$$y_{ij} \in B_{\delta_{ij}'}(y_{ij}) \subset B_{\delta_{ij}}(x_{ij}).$$
(56)

It follows from (11), (13), and (56) that for $1 \le i \le 2$, $1 \le j \le m_i - 1$,

$$\|F(x) - F(y)\| \ge \|f(x) - f(y)\| - \|g(x) - g(y)\| \ge (\lambda_{i0} - L)\|x - y\|, \quad \forall x, y \in \overline{B}_{r_i^*}(z_i^*),$$
(57)

$$\|F(x) - F(y)\| \ge \|f(x) - f(y)\| - \|g(x) - g(y)\| \ge (\mu_{ij} - L)\|x - y\|, \quad \forall x, y \in \overline{B}_{\delta'_{ij}}(y_{ij}),$$
(58)

where $\lambda_{i0} > L + 1$ and $\mu_{ij} > L$ can be derived from (45), (46), (53), and (54).

For each $i(1 \le i \le 2)$, since z_i^* is a regular expanding fixed point of F and $F(B_{r_i^*}(z_i^*))$ is an open set, the backward orbit of y_{i0} lies in $B_{r_i^*}(z_i^*)$ and (57) holds in some neighborhood of each point on the backward orbit. The forward orbit of y_{i0} consists of y_{ij} for $1 \le j \le m_i - 1$ and (58) holds for each y_{ij} in $\overline{B}_{\delta'_{ij}}(y_{ij})$. Therefore, the heteroclinic cycle Γ' connecting repellers z_1^* and z_2^* is nondegenerate. In addition, it is clear that F is continuous in $B_{r_i^*}(z_i^*)$ and $B_{\delta'_{ij}}(y_{ij})$ for $1 \le i \le 2$, $1 \le j \le m_i - 1$. It follows from (3) of Remark 2.2 in [25] that if we prove that for each point y_0 on the cycle Γ' , there exists a positive constant r_0 such that $F(y_0)$ is an interior point of $F(B_{r_0}(y_0))$, then this cycle Γ' is regular. Firstly, for each point y_0 on Γ' lying in $B_{r_i^*}(z_i^*)$, there

Firstly, for each point y_0 on Γ' lying in $B_{r_i^*}(z_i^*)$, there exists a constant r_0 such that $\overline{B}_{r_0}(y_0) \in B_{r_i^*}(z_i^*)$. It follows from (57) and Lemma 1, by using F to replace f and making g = 0, that

$$B_{(\lambda_{i0}-L)r_0}(F(y_0)) \in F(B_{r_0}(y_0)),$$
(59)

which implies that $F(y_0)$ is an interior point of $F(B_{r_0}(y_0))$.

Secondly, for each point y_{ij} , $1 \le j \le m_i - 1$, on Γ' lying out $B_{r_i^*}(z_i^*)$, it follows from (58) and Lemma 1, by using *F* to replace *f* and making q = 0 again, that

$$B_{(\mu_{ij}-L)\delta'_{ij}}(F(y_{ij})) \subset F(B_{\delta'_{ij}}(y_{ij})),$$
(60)

which implies that $F(y_{ij})$ is an interior point of $F(B_{\delta'_{ij}}(y_{ij}))$. Hence, the cycle Γ' is regular. That is, the heteroclinic

Hence, the cycle Γ' is regular. That is, the heteroclinic cycle Γ' connecting repellers z_1^* and z_2^* of F is regular and nondegenerate. Consequently, it follows form Lemma 2 that there exists an uncountable, perfect, bounded, and closed set V such that system (9) is chaotic on V in the sense of both Devaney and Li-Yorke. This completes the proof.

Remark 4. Theorem 1 gives a relatively explicit range of the Lipschitz perturbation *g* which is characterized by a constant

 ε_0 determined by the properties of the original map f. From (10), we see that it only needs L and the values of g at z_i , x_{ij} are less than ε_0 , and it does not need to compute all the values of g in some domains. Hence, the conditions about g in Theorem 1 are relatively easy to check out in practice. In addition, it only needs the original map f to be continuous near some points of interest without having to be continuous in the whole space.

Remark 5. From the proof of Theorem 1, it is easy to see that the perturbed map *F* will have a regular and nondegenerate heteroclinic cycle Γ' connecting repellers if the unperturbed map *f* with a regular and nondegenerate heteroclinic cycle Γ connecting repellers undergoes a small perturbation, and the cycle Γ' is near to Γ . The perturbed range of *g* is characterized by ε_0 determined in Theorem 1. Thus, this result can be viewed as persistence of regular and nondegenerate heteroclinic cycles connecting repellers in Banach spaces.

When the original map f is continuously differentiable in some domains of interest, using a similar method to Theorem 1, we can get the following result.

Theorem 2. Let $(X, \|\cdot\|)$ be a Banach space and $f: X \longrightarrow X$ be a map with $k (\geq 2)$ different fixed points $z_1, \ldots, z_k \in X$. Assume that

- (i) For each $i(1 \le i \le k)$, f is continuously differentiable in $B_{r'_i}(z_i)$ for some constant $r'_i > 0$ and $Df(z_i)$ is an invertible linear map satisfying $||Df(z_i)||^0 > 1$, which is equivalent to that there exists a positive constant $r_i \le r'_i$ such that z_i is a regular expanding fixed point of f in $\overline{B}_{r_i}(z_i)$.
- (ii) f has a heteroclinic cycle Γ connecting repellers z_1, \ldots, z_k .
- (iii) f is continuously differentiable in some neighborhood U_{x_0} of each point x_0 on the cycle Γ , and $Df(x_0)$ is an invertible linear map.

Then, there exists a constant $\varepsilon_0 > 0$ such that for any Lipschitz map g in each set of $\overline{B}_{r_i}(z_i)$ and U_{x_0} for $x_0 \in \Gamma$, with the Lipschitz constant L satisfying

$$\max\{L, \|g(x_0)\| \text{ for } x_0 \in \Gamma\} < \varepsilon_0, \tag{61}$$

the results of Theorem 1 hold.

Proof. It follows from the assumptions in Theorem 2 and Lemma 3 that f has a regular and nondegenerate heteroclinic cycle Γ connecting repellers z_1, \ldots, z_k . For each $i(1 \le i \le k)$, since z_i is a regular expanding fixed point of f, there exist a point $x_{i0} \in B_{r_i}(z_i)$ and a positive integer $m_i \ge 1$ such that $f(x_{i0}) \notin B_{r_i}(z_i)$ and $f^{m_i}(x_{i0}) = z_{t(i)}$, where $t(i) = [i \mod k] + 1$. The rest of the proof is similar to that of Theorem 1, so it is omitted.

For a function $f \in C^1(U, X)$, the following norm is often used

$$\|f\|_{C^{1},U} := \sup\{\|f(x)\|, \|Df(x)\|, x \in U \subset X\}.$$
(62)

Therefore, if the conditions in Theorems 1 and 2 about g are replaced by those based on the above norm, then we can obtain two consequences of Theorems 1 and 2. For convenience, we list them as the following theorems.

Theorem 3. Suppose that $(X, \|\cdot\|)$ is a Banach space, $f: X \longrightarrow X$ is a map with $k (\ge 2)$ different fixed points z_1 , $\ldots, z_k \in X$ and satisfies the conditions (i) and (ii) in Theorem 1. Then, there exists a constant $\varepsilon_0 > 0$ such that for any $g \in C^1(U, X)$ with $\|g\|_{C_1^1U} < \varepsilon_0$, the results of Theorem 1 hold, where $U = B_{r_i}(z_i) \cup (\bigcup_{j=1}^{m_i-1} U_{ij}), 1 \le i \le k$.

Theorem 4. Suppose that $(X, \|\cdot\|)$ is a Banach space, $f: X \longrightarrow X$ is a map with $k (\geq 2)$ different fixed points $z_1, \ldots, z_k \in X$ and satisfies the conditions (i)–(iii) in Theorem 2. Then, there exists a constant $\varepsilon_0 > 0$ such that for any $g \in C^1(U, X)$ with $\|g\|_{C^1, U} < \varepsilon_0$, the results of Theorem 1 hold, where $U = \bigcup_{x_0 \in I} U_{x_0}$.

At the last of this section, we discuss a usually used Banach space \mathbb{R}^n , which is the Euclidean space. As is well known, there are many different norms in \mathbb{R}^n . A map in \mathbb{R}^n can expand in different norms, see [11, 26] and references therein. It is natural to ask whether there is the persistence of heteroclinic cycles connecting repellers in \mathbb{R}^n , where the repellers expand in different norms. The following Theorem 5 will answer this question.

The usually used Euclidean norm is denoted by

$$\|x\| = \left(\sum_{j=1}^{n} |x_j|^2\right)^{1/2}, \quad x = (x_1, \dots, x_n)^T \in \mathbf{R}^n.$$
 (63)

In the following, we will use the neighborhood of a point $x \in \mathbf{R}^n$ in different norms. For convenience, let $\overline{B}_r(x)$ and $B_r(x)$ denote the closed and open balls of x with radius r in

Now, we establish a result on persistence of heteroclinic cycles connecting repellers in \mathbb{R}^n , where the repellers expand in different norms.

Theorem 5. Suppose that a map $f: \mathbb{R}^n \longrightarrow \mathbb{R}^n$ has $k (\geq 2)$ different fixed points $z_1, \ldots, z_k \in \mathbb{R}^n$ and satisfies the following conditions:

- (i) for each $i(1 \le i \le k)$, f is continuously differentiable in some neighborhood of z_i and all the eigenvalues of $Df(z_i)$ have absolute values larger than 1, which implies that there exist a constant $r_i > 0$ and a norm $\| \cdot \|_i$ in \mathbb{R}^n such that f is continuously differentiable in $\overline{N}_{r_i}(z_i)$, and z_i is a regular expanding fixed point of fin $\overline{N}_{r_i}(z_i)$.
- (ii) for each $i(1 \le i \le k)$, there exist a point $x_{i0} \in N_{r_i}(z_i)$, $x_{i0} \ne z_i$, and a positive integer $m_i \ge 1$ such that $f^{m_i}(x_{i0}) = z_{t(i)}$, where $t(i) = [i \mod k] + 1$. Furthermore, f is continuously differentiable in some neighborhood U_{ij} of x_{ij} and satisfies that $detDf(x_{ij}) \ne 0$, where $x_{ij} = f^j(x_{i0})$ for $1 \le j \le m_i - 1$.

Then, for any Lipschitz map g with Lipschitz constant L in the Euclidean norm $\|\cdot\|$ in each set of $\overline{N}_{r_i}(z_i)$ and U_{ij} , $1 \le i \le k, 1 \le j \le m_i - 1$, there exists a constant $\varepsilon_0 > 0$ satisfying

$$\max\left\{L, \|g(z_i)\|, \|g(x_{ij})\|, \quad 1 \le i \le k, \ 0 \le j \le m_i - 1\right\} < \varepsilon_0,$$
(64)

such that the perturbed system (9) is chaotic in the sense of both Devaney and Li-Yorke on a compact and perfect set which contains a Cantor set.

Proof. Without loss of generality and for simplicity, we also only show that Theorem 5 holds for k = 2.

For convenience, let i = 1 or 2 in the rest of proof. As pointed in the second paragraph of the proof in Theorem 1, we can also suppose that $N_{r_1}(z_1) \cap N_{r_2}(z_2) = \emptyset$ and $f(x_{i0}) \notin N_{r_i}(z_i)$.

Since all the norms on \mathbb{R}^n are equivalent by Corollary 3.14 of Chapter II in [36], there exist positive constants b_{11} , b_{12} , c_{i1} and c_{i2} such that

$$b_{11} \| \cdot \|_{1} \le \| \cdot \|_{2} \le b_{12} \| \cdot \|_{1},$$

$$c_{i1} \| \cdot \|_{i} \le \| \cdot \| \le c_{i2} \| \cdot \|_{i}.$$
(65)

Since *g* is a Lipschitz map with Lipschitz constant *L* in the Euclidean norm $\|\cdot\|$ in $\overline{N}_{r_i}(z_i)$ and U_{ij} , for any $x, y \in \overline{N}_{r_i}(z_i)$ and any $x, y \in U_{ij}$, $1 \le j \le m_i - 1$, it follows from (65) that

$$\|g(x) - g(y)\|_{i} \le c_{i1}^{-1} \|g(x) - g(y)\| \le c_{i1}^{-1} L \|x - y\| \le c_{i1}^{-1} c_{i2} L \|x - y\|_{i} \le L' \|x - y\|_{i},$$
(66)

where

$$L' = \max\{c_{i1}^{-1}c_{i2}L, \quad i = 1, 2\}.$$
(67)

Then it follows from (66) that g is also a Lipschitz map with Lipschitz constant L' in the norm $\|\cdot\|_i$ in $\overline{N}_{r_i}(z_i)$ and U_{ij} .

It follows from assumption (i) that there exists a constant $\lambda_{i0} > 1$ such that

$$\|f(x) - f(y)\|_i \ge \lambda_{i0} \|x - y\|_i, \quad \forall x, y \in \overline{N}_{r_i}(z_i),$$
 (68)

 $f: \overline{N}_{r_i}(z_i) \longrightarrow f(\overline{N}_{r_i}(z_i))$ is a homeomorphism and $f(N_{r_i}(z_i))$ is open, f(D) is open for any open set $D \in N_{r_i}(z_i)$. Take a constant

$$\delta_{i0} < \frac{r_i - \|z_i - x_{i0}\|_i}{2},\tag{69}$$

such that $\overline{N}_{\delta_{i0}}(x_{i0}) \subset N_{r_i}(z_i)$. Then, it follows form (68) that $f: N_{\delta_{i0}}(x_{i0}) \longrightarrow f(N_{\delta_{i0}}(x_{i0}))$ is also a homeomorphism. In addition, it follows from $\det Df(x_{ij}) \neq 0$,

In addition, it follows from det $Df(x_{ij}) \neq 0$, $1 \leq j \leq m_i - 1$, that none of the eigenvalues of $Df(x_{ij})$ is 0. Therefore, $(Df(x_{ij}))^T Df(x_{ij})$ is positive definite. Then,

$$\| Df(x_{ij})\|^{0} = \left(\inf_{\|x\|=1} \left(x^{T} \left(Df(x_{ij}) \right)^{T} Df(x_{ij}) x \right) \right)^{\frac{1}{2}} > 0,$$
(70)

where $x \in \mathbf{R}^n$. It follows from (65) and (70) that

$$\begin{split} \left\| Df(x_{ij}) \right\|_{i}^{0} &= \inf_{x \neq 0} \frac{\left\| Df(x_{ij})x \right\|_{i}}{\|x\|_{i}} \ge c_{i1}c_{i2}^{-1}\inf_{x \neq 0} \frac{\left\| Df(x_{ij})x \right\|}{\|x\|} \\ &= c_{i1}c_{i2}^{-1} \left\| Df(x_{ij}) \right\|^{0} > 0, \end{split}$$
(71)

Hence, it follows from (71) and Lemma 4 that there exist positive constants μ_{ij} and δ_{ij} such that

$$\|f(x) - f(y)\|_i \ge \mu_{ij} \|x - y\|_i, \quad \forall x, y \in \overline{N}_{\delta_{ij}}(x_{ij}), \tag{72}$$

which implies that $f: N_{\delta_{ij}}(x_{ij}) \longrightarrow f(N_{\delta_{ij}}(x_{ij}))$ is homeomorphic, and $f(N_{\delta_{ij}}(x_{ij}))$ is open for $1 \le j \le m_i - 1$, where δ_{ij} satisfies the following conditions:

$$\delta_{i1} < \lambda_{i0} \delta_{i0},$$

$$\delta_{i,j+1} < \mu_{ij} \delta_{ij}, \quad \text{for } 1 \le j \le m_i - 2,$$
(73)

 $\overline{N}_{\delta_{ij}}(x_{ij})$ are disjoint subsets of U_{ij} and $\overline{N}_{\delta_{ij}}(x_{ij}) \cap N_{r_i}(z_i) = \emptyset$ for fixed *i* and $1 \le j \le m_i - 1$.

The rest of the proof is almost exactly the same to Steps 1-3 in the proof of Theorem 1 except for three aspects. One is that *L* is replaced by *L'* and the domains in the norm $\|\cdot\|$ are replaced by those in the norms $\|\cdot\|_1$ or $\|\cdot\|_2$, respectively. In brief, in the representations of the domains, the alphabet B is replaced by the alphabet N through the proof of Theorem 1. The second is that some values in the norm $\|\cdot\|$ are replaced by those in the norms $\|\cdot\|_1$ or $\|\cdot\|_2$, respectively. It is pointed out that (18) and (42) take the values in the norm $\|\cdot\|_1$, the remainders follow the following rule: if the independent variables of functions are taken from $N_r(z_i)$ or $N_{\delta_{ii}}(x_{ij})$, then the values in the norm $\|\cdot\|$ are replaced by those in the norm $\|\cdot\|_i$. The third is that some related constants used in the proof are slightly modified since the norm $\|\cdot\|$ is replaced by the norms $\|\cdot\|_1$ or $\|\cdot\|_2$. For convenience, we list them as follows.

$$\delta_{1,m_{1}} < \begin{cases} \min \left\{ \begin{array}{l} b_{11}\lambda_{10}\delta_{10}, \\ \frac{r_{2} - \|z_{2} - x_{20}\|_{2}}{2} - \delta_{20} \end{array} \right\}, & \text{if } m_{1} = 1; \\ \min \left\{ \begin{array}{l} \frac{b_{11}\mu_{1,m_{1}-1}\delta_{1,m_{1}-1}, \\ \frac{r_{2} - \|z_{2} - x_{20}\|_{2}}{2} - \delta_{20} \end{array} \right\}, & \text{if } m_{1} > 1, \\ \frac{r_{2} - \|z_{2} - x_{20}\|_{2}}{2} - \delta_{20} \end{array} \right\}, & \text{if } m_{1} > 1, \\ \delta_{2,m_{2}} < \begin{cases} \min \left\{ \begin{array}{l} \frac{b_{11}^{-1}\lambda_{20}\delta_{20}, \\ \frac{r_{1} - \|z_{1} - x_{10}\|_{1}}{2} - \delta_{10} \end{array} \right\}, & \text{if } m_{2} = 1; \\ \min \left\{ \begin{array}{l} \frac{b_{12}^{-1}\mu_{2,m_{2}-1}\delta_{2,m_{2}-1}, \\ \frac{r_{1} - \|z_{1} - x_{10}\|_{1}}{2} - \delta_{10} \end{array} \right\}, & \text{if } m_{2} > 1, \end{cases}$$

$$r_{1}^{*} = \frac{r_{1} + \left\|z_{1} - x_{10}\right\|_{1}}{2},$$

$$r_{2}^{*} = \frac{r_{1} + \left\|z_{2} - x_{20}\right\|_{2}}{2},$$

$$\varepsilon_{1} = \frac{(\lambda_{10} - 1)\delta_{2,m_{2}}}{1 + \delta_{2,m_{2}}},$$

$$\varepsilon_{2} = \frac{(\lambda_{20} - 1)\delta_{1,m_{1}}}{1 + \delta_{1,m_{1}}},$$

$$\varepsilon_{3} = \min\left\{\varepsilon_{1}, \varepsilon_{2}, \frac{\min\{\lambda_{10}\delta_{10} - \delta_{11}, \mu_{1,m_{1}-1}\delta_{1,m_{1}-1} - b_{11}^{-1}\delta_{1,m_{1}}, \mu_{1j}\delta_{1j} - \delta_{1,j+1}, \text{ for } 1 \le j \le m_{1} - 2\}}{1 + \max\{\delta_{1j}, 0 \le j \le m_{1} - 1\}}\right\},$$

$$\varepsilon_{4} = \min\left\{\varepsilon_{1}, \varepsilon_{2}, \frac{\min\{\lambda_{20}\delta_{20} - \delta_{21}, \mu_{2,m_{2}-1}\delta_{2,m_{2}-1} - b_{12}\delta_{2,m_{2}}, \mu_{2j}\delta_{2j} - \delta_{2,j+1}, \text{ for } 1 \le j \le m_{2} - 2\}}{1 + \max\{\delta_{2j}, 0 \le j \le m_{2} - 1\}}\right\}.$$
(74)

It follows from (73) and the above conditions that $\varepsilon_j > 0$ for $1 \le j \le 4$. Here, it should explain how to take the values of the numerators in terms of m_1 and m_2 in ε_3 and ε_4 . We only explain the values in terms of m_1 , while that for m_2 is similar. For the numerator of the fraction at the right side of ε_3 , when $m_1 = 1$, it only takes the first term; when $m_1 = 2$, it only takes the first two terms; when $m_1 \ge 3$, it takes all of the terms. Set

$$\varepsilon_{0}' = \min\left\{\varepsilon_{j}, \ 1 \le j \le 4\right\}, \quad \varepsilon_{0} = \min\left\{\frac{\varepsilon_{0}'}{\max\left\{c_{11}^{-1}c_{12}, c_{21}^{-1}c_{22}\right\}}, \ c_{11}\varepsilon_{0}', \ c_{21}\varepsilon_{0}'\right\}.$$
(75)

If *g* satisfies condition (64) in the Euclidean norm $\|\cdot\|$, that is the following:

$$\max\left\{L, \|g(z_i)\|, \|g(x_{ij})\|, 1 \le i \le 2, 0 \le j \le m_i - 1\right\} < \varepsilon_0,$$
(76)

then it follows from (65), (75) and (76) that *g* also satisfies the following condition in the norms $\|\cdot\|_1$ and $\|\cdot\|_2$

$$\max\left\{L', \left\|g(z_{i})\right\|_{i}, \left\|g(x_{ij})\right\|_{i}, 1 \le i \le 2, 0 \le j \le m_{i} - 1\right\} < \varepsilon_{0}',$$
(77)

Therefore, repeating the Steps 1–3 in the proof of Theorem 1, we can get that if g satisfies (76), consequently (77), then F has a regular and nondegenerate heteroclinic cycle Γ' connecting repellers z_1^* and z_2^* in different norms $\|\cdot\|_1$ and $\|\cdot\|_2$. It follows from the proof of Theorem 4.1 in [26] that there exists a positive integer p such that F^p has a heteroclinic cycle Γ^* connecting repellers z_1^* and z_2^* in the unified Euclidean norm $\|\cdot\|$. Since all the points on the cycle Γ^* of F^p also lie on the cycle Γ' of F, it is easy to prove that the cycle Γ^* of F^p is also regular and nondegenerate. Consequently, F is chaotic on a compact and perfect set which contains a Cantor set in the sense of both Devaney and Li-Yorke. This completes the proof.

Remark 6. From the proof of Theorem 5, we obtain that F^p has a regular and nondegenerate heteroclinic cycle

connecting repellers in the unified Euclidean norm $\|\cdot\|$ for some positive integer p. Hence, Theorem 5 can also be regarded as the persistence of a regular and nondegenerate heteroclinic cycle connecting repellers in \mathbb{R}^n . In the special case that all the norms $\|\cdot\|_i$, $1 \le i \le k$, in assumption (i) become a unified norm, such as the Euclidean norm $\|\cdot\|$, then the positive integer p becomes 1. Hence, this special case of Theorem 5 is consistent with Theorem 1.

The following result is a direct consequence of Theorem 5.

Theorem 6. Suppose that a map $f: \mathbb{R}^n \longrightarrow \mathbb{R}^n$ has $k (\geq 2)$ different fixed points $z_1, \ldots, z_k \in \mathbb{R}^n$ and satisfies the following conditions

- (i) For each i (1 ≤ i ≤ k), z_i is an expanding fixed point of f in some norm || · ||_i;
- (ii) f has a k-heteroclinic cycle Γ connecting fixed points z₁,..., z_k and is continuously differentiable in some neighborhood U_{x₀} of each point x₀ ∈ Γ satisfying detDf(x₀) ≠ 0.

Then, there exists a constant $\varepsilon_0 > 0$ such that for any Lipschitz map g in each set of $\bigcup_{x_0 \in \Gamma} U_{x_0}$ with Lipchitz constant L in the Euclidean norm $\|\cdot\|$ satisfying

$$\max\{L, \|g(x_0)\|, \quad \text{for } x_0 \in \Gamma\} < \varepsilon_0, \tag{78}$$

the results in Theorem 5 hold.

Remark 7. [28] studied the persistence of heteroclinic repellers in \mathbb{R}^n for C^1 maps with C^1 perturbations, where the maps needed to be continuously differentiable in the whole space. Here, it only needs the maps to be continuously differentiable in some neighborhoods of points. The main differences between the above two theorems and the result in [28] are as follows. One is that Theorems 5 and 6 studied the Lipschitz perturbations, while the latter considered the C^1 perturbations. The second is that Theorems 5 and 6 give an explicit expression for the range of perturbations, while the latter did not give such a range for perturbations. The third is that Theorems 5 and 6 use different norms for expansions of fixed points which are more general in practice, while the latter only used a single norm for expansions of fixed points.

Remark 8. Just as Theorems 3 and 4, if the perturbed term g is continuously differentiable, then the conditions about g in Theorems 5 or 6 can be replaced by that $g \in C^1(U, X)$ with $\|g\|_{C^1,U} < \varepsilon_0$, where U is taken the corresponding domains used in Theorems 5 or 6, respectively, then the results in Theorems 5 or 6 hold.

4. Examples

In this section, three examples are given to illustrate the validity of the theoretical results.

Example 3. The original map f is taken as the following map on **R**:

$$f(x) = \begin{cases} 2x, & \text{if } x \in [-2, 2] \\ 5x - 9, & \text{if } x \in (2, 2.5) \\ 0.1x - 0.2875, & \text{else.} \end{cases}$$
(79)

The perturbed map g is taken as $g(x) = \gamma |x|$, where $x \in \mathbf{R}$ and γ is a positive real number. It is obvious that f is piecewise continuous on \mathbf{R} , g is a Lipschitz map with a Lipschitz constant $L = \gamma$ and it is not differentiable on \mathbf{R} .

It is easy to see that $z_1 = 0$ and $z_2 = 2.25$ are two regular expanding fixed points of *f*. Set $x_{10} = 1.125 \in (-2, 2)$, then

 $f(x_{10}) = z_2$. Set $x_{20} = 2.375 \in (2, 2.5)$, then $x_{21} = f(x_{20}) = 2.875$ and $f(x_{21}) = z_1$, that is, $f^2(x_{20}) = z_1$. So, f has a 2-heteroclinic cycle Γ connecting repellers z_1 and z_2 . It is clear that the cycle is regular and nondegenerate, and assumptions (i) and (ii) in Theorem 1 holds with k = 2, $r_1 = 2$, $r_2 = 0.25$, $\lambda_{10} = 2$, $\lambda_{20} = 5$, $m_1 = 1$, $m_2 = 2$, x_{10} and x_{20} as the above. Some constants that appear in the proof of Theorem 1

are taken as follows: $\mu_{21} = 0.1$, $\delta_{10} = 0.43 < (r_1 - |z_1|)$ $\begin{array}{l} -x_{10}|)/2) = 0.4375, \quad \delta_{20} = 0.06 < \quad ((r_2 - |z_2 - x_{20}|)/2) \\ = 0.0625, \quad \delta_{11} = 0.002 < \min\{\lambda_{10}\delta_{10}, \quad ((r_2 - |z_2 - x_{20}|)/2) \\ -\delta_{20}\} = 0.0025, \quad \delta_{21} = 0.28 < \lambda_{20}\delta_{20} = 0.3, \quad \delta_{22} = 0.007 < 0.000 < 0.0000 \\ \end{array}$ $\min\{\mu_{21}\delta_{21}, ((r_1 - |z_1 - x_{10}|)/2) - \delta_{10}\} = 0.0075, \ \varepsilon_1 = (\lambda_{10})$ $\approx 0.006951, \qquad \varepsilon_2 = (((\lambda_{20} - 1)\delta_{11})/1 +$ $(-1)\delta_{22}/1 + \delta_{22})$ $\delta_{11} \approx 0.007984, \ \ \varepsilon_3 = \min\{\varepsilon_1, \varepsilon_2, ((\delta_{10}\lambda_{10} - \delta_{11})/1 + \delta_{10})\}$ = 0.006951, $\varepsilon_4 = \min \{\varepsilon_1, \varepsilon_2, (\min \{\lambda_{20}\delta_{20} - \delta_{21}, \mu_{21}\delta_{21} - \delta_{22}\})/1 + \max\{\delta_{20}, \delta_{21}\}\} = 0.006951, \varepsilon_0 = \min$ $|\varepsilon_j, 1 \le j \le 4| = 0.006951$. It is easy to check that the perturbation g satisfies condition (10) for $\gamma \leq 0.0024$. Then, it follows from the result of Theorem 1 that the perturbed system F = f + g also has a regular and nondegenerate heteroclinic cycle Γ' connecting repellers which is near to Γ . Consequently, F and f are chaotic in the sense of both Devaney and Li-Yorke. For illustrating the persistence of a heteroclinic cycle connecting repellers, we take $\gamma = 0.002$ for example. It is easy to calculate the following results. The perturbed map F has two regular expanding fixed pints $z_1^* =$ 0 and $z_2^* = (1500/667) \approx 2.248876$. There exist two points $x_{10}^* = (750000/667667) \approx 1.123314$ and $x_{20}^* = (301375)$ $(127551) \approx 2.362780$ such that $F(x_{10}^*) = z_2^*, x_{21}^* = F(x_{20}^*) =$ $(575/204) \approx 2.818627$ and $F(x_{21}^*) = z_1^*$. Then, F has a 2heteroclinic cycle Γ' connecting repellers $z_1^* = 0$ and $z_2^* = 0$. It is clear that Γ' is near to Γ . With the increase of γ , the heteroclinic cycle Γ' will gradually run away from Γ until it breaks or disappears. Since f and F are chaotic on some intervals of **R** and the computer simulations of them are on intervals, we omit the computer simulations.

Example 4. The original map f is taken as the following map on \mathbb{R}^2

$$f(x, y) = \begin{cases} 8(x, y), & \text{if } (x, y) \in \overline{B}_{1}(0, 0), \\ (2x - 2, 2y - 2), & \text{if } (x, y) \in \overline{B}_{4}(0, 0)/\overline{B}_{1}(0, 0), \\ \left(\sin \left[x - 2 - \frac{\pi}{2} + \left(y - 2 - \frac{\pi}{2} \right)^{2} \right], \\ \sin \left[\left(x - 2 - \frac{\pi}{2} \right)^{2} + y - 2 - \frac{\pi}{2} \right] \right), & \text{if } (x, y) \notin \overline{B}_{4}(0, 0). \end{cases}$$

$$(80)$$

This map is used as an example in [25] for illustrating chaos induced by a heteroclinic cycle connecting repellers.

The perturbed map *g* is taken as $g(x, y) = \gamma(x, y)$, where $(x, y) \in \mathbb{R}^2$ and γ is a real number. It is obvious that *f* is only

continuously differentiable in some domains of \mathbf{R}^2 , g is continuously differentiable in \mathbf{R}^2 and has a Lipschitz constant $L = |\gamma|$.

On the one hand, it is clear that $z_1 = (0,0)$ and $z_2 = (2,2)$ are two fixed points of f, f is continuously differentiable in $\overline{B}_1(z_1)$, $\overline{B}_{1.171}(z_2)$, and satisfies that

$$Df(z_1) = 8I_2,$$

 $Df(z_2) = 2I_2,$ (81)

where I_2 is the identity matrix. So, the eigenvalues of $Df(z_1)$ and $Df(z_2)$ have absolute values larger than 1, which implies that z_1 and z_2 are two regular expanding expanding fixed points of f in $\overline{B}(z_1)$ and $\overline{B}_{1.171}(z_2)$ in the Euclidean norm $\|\cdot\|$ with $\lambda_{10} = 8$, $\lambda_{20} = 2$, respectively. It is obvious that $\overline{B}_1(z_1) \cap \overline{B}_{1.171}(z_2) = \emptyset$ and both lie in $B_4(z_1)$. Set $x_{10} = ((1/4), (1/4)) \in B_1(z_1)$, then $f(x_{10}) = z_2$. Set $x_{20} = (2 + (\pi/8), 2 + (\pi/8)) \in B_{1.171}(z_2)$, then $x_{21} =$ $f(x_{20}) = (2 + (\pi/4), 2 + (\pi/4)) \in B_4(z_1) \setminus B_1(z_1)$, $x_{22} =$ $f(x_{21}) = (2 + (\pi/2), 2 + (\pi/2)) \notin \overline{B}_4(z_1)$ and $f(x_{22}) = z_1$, that is, $f^3(x_{20}) = z_1$.

On the other hand, it is also obvious that f is continuously differentiable in some neighborhoods of x_{10} , x_{20} , x_{21} , and x_{22} and satisfies

$$Df(x_{10}) = 8I_2,$$

$$Df(x_{20}) = Df(x_{21})$$

$$= 2I_2,$$

$$Df(x_{22}) = I_2.$$

(82)

Then, it follows from (82) that

$$\|Df(x_{10})\|^{0} = 8,$$

$$\|Df(x_{20})\|^{0} = \|Df(x_{21})\|^{0}$$

$$= 2,$$

$$\|Df(x_{22})\|^{0} = 1,$$

(83)

which together with Lemma 4 imply that the cycle Γ is nondegenerate. Furthermore, it follows from (4) of Remark 2.2 in [25] that the cycle Γ is also regular. Consequently, f has a regular and nondegenerate 2-heteroclinic cycle Γ connecting the repellers z_1 and z_2 .

Therefore, assumptions (i) and (ii) in Theorem 5 hold with k = 2, $r_1 = 1$, $r_2 = 1.171$, $\lambda_{10} = 8$, $\lambda_{20} = 2$, $m_1 = 1$, $m_2 = 3$, x_{10} and x_{20} as the above. Consequently, f has a 2heteroclinic cycle Γ connecting repellers z_1 and z_2 . As is pointed out in Remark 6, when the norms used in Theorem 5 become a unified norm, the special case of Theorem 5 is consistent with Theorem 1. So, we can take some constants that appear in the proof of Theorem 1 as follows:

$$\begin{split} \mu_{21} &= 2, \\ \mu_{22} &= 1, \\ \delta_{10} &= 0.2 < \frac{r_1 - \left\| z_1 - x_{10} \right\|}{2} \approx 0.323223, \\ \delta_{20} &= 0.175 < \frac{r_2 - \left\| z_2 - x_{20} \right\|}{2} \approx 0.307820, \\ \delta_{11} &= 0.1328 < \min\left\{ \lambda_{10} \delta_{10}, \frac{r_2 - \left\| z_2 - x_{20} \right\|}{2} - \delta_{20} \right\} \\ &= 0.132820, \\ \delta_{21} &= 0.2 < \lambda_{20} \delta_{20} \\ &= 0.35, \\ \delta_{22} &= 0.2 < \mu_{21} \delta_{21} \\ &= 0.4, \\ \delta_{23} &= 0.025 < \min\left\{ \mu_{22} \delta_{22}, \frac{r_1 - \left\| z_1 - x_{10} \right\|}{2} - \delta_{10} \right\} \\ &= 0.123223, \\ \epsilon_1 &= \frac{(\lambda_{10} - 1)\delta_{23}}{1 + \delta_{23}} \approx 0.170732, \\ \epsilon_2 &= \frac{(\lambda_{20} - 1)\delta_{11}}{1 + \delta_{11}} \approx 0.117232, \\ \epsilon_3 &= \min\left\{ \epsilon_1, \epsilon_2, \frac{\delta_{10}\lambda_{10} - \delta_{11}}{1 + \delta_{10}} \right\} \\ &= 0.117232, \\ \epsilon_4 &= \min\left\{ \epsilon_1, \epsilon_2, \frac{\min\{\lambda_{20}\delta_{20} - \delta_{21}, \mu_{21}\delta_{21} - \delta_{22}, \mu_{22}\delta_{22} - \delta_{23}\} \right\} \\ &= 0.117232, \end{split}$$
(84)

 $\varepsilon_0 = \min\{\varepsilon_j, 1 \le j \le 4\} = 0.117232$. It is easy to check that the perturbation *g* satisfies condition (10) for $|\gamma| \le 0.0297$. Then, it follows from the result of Theorem 5 that the perturbed system F = f + g also has a regular and nondegenerate heteroclinic cycle Γ' connecting repellers which is near to Γ . Consequently, *F* and *f* are chaotic in the sense of both Devaney and Li-Yorke.

As is done in Example 3, for a given γ , one can also directly calculate the heteroclinic cycle Γ' of *F* to check whether it is near to Γ of *f*. However, it is not easy to directly calculate such a cycle for high-dimensional maps. If there is

the persistence of a heteroclinic cycle connecting repellers, then the computer simulations of them will not change very much. The behaviors of the unperturbed map f with an initial point (x, y) = (0.1, 0.1) are illustrated in Figure 1. We do some computers simulations of F as γ increases from -0.0297 to 0 or from 0 to 0.0297, and find that all the simulations are similar with that of the original map f in Figure 1. Here, we give one simulation of F with an initial point (x, y) = (0.1, 0.1) for y = 0.0297, see Figure 2. We can see that Figure 2 is a small change to Figure 1, which shows that the heteroclinic cycle Γ' of *F* is near to Γ of *f*. When we let |y| continuous to increase, we find that the computer simulations gradually change until there is a big difference from that of the original map. This shows that the heteroclinic cycle Γ breaks or disappears. Are there new heteroclinic cycles connecting repellers not near Γ or new snapback repellers to make the perturbed system still chaotic? It is an interesting question, while it is out of the scope of this paper and will be our further study.

Example 5. The original map f is taken as the following map on \mathbb{R}^3

$$f(x, y, z) = \begin{cases} (6x, 6y, 6z), & \text{if } (x, y, z) \in \overline{B}_1(O), \\ (4x - 9, 4y - 9, 4z - 9), & \text{if } (x, y, z) \in \overline{B}_8(O)/\overline{B}_1(O), \\ (\sin[x - 5 + (y - 5)^2], & \sin[y - 5 + (z - 5)^2], \\ \sin[(x - 5)^2 + z - 5]), & \text{if } (x, y, z) \notin \overline{B}_8(O), \end{cases}$$
(85)

where O = (0, 0, 0) is the origin. The perturbed map g is taken as $g(x, y, z) = \gamma(x, y, z)$, where $(x, y, z) \in \mathbb{R}^3$ and γ is a real number. It is obvious that f is only continuously differentiable in some domains of \mathbb{R}^3 , g is continuously differentiable in \mathbb{R}^3 and has a Lipschitz constant $L = |\gamma|$.

Theorem 5 is also used to verify the persistence of a heteroclinic cycle connecting repellers, and the process is similar to that of Example 4. So, we omit some details and only give some main results as follows. Assumptions (i) and (ii) in Theorem 5 hold with k = 2, $z_1 = (0, 0, 0)$, $z_2 = (3, 3, 3)$, $r_1 = 1$, $r_2 = 2.8$, $\lambda_{10} = 6$, $\lambda_{20} = 4$, $m_1 = 1$, $m_2 = 2$, $x_{10} = (0.5, 0.5, 0.5) \in B_1(z_1)$ and $x_{20} = (3.5, 3.5, 3.5) \in B_{2.8}(z_2) \subset B_8(O)$. The points z_1 and z_2 are two regular expanding fixed points of f in the Euclidean norm $\|\cdot\|$. In addition, $f(x_{10}) = z_2$, $x_{21} = f(x_{20}) = (5, 5, 5) \notin \overline{B}_8(O)$ and $f(x_{21}) = z_1$, that is, $f^2(x_{20}) = z_1$. Then f has a regular and nondegenerate 2-heteroclinic cycle



FIGURE 1: Complex behaviors of the original map f in the (x, y) space, where the initial point is take as (0.1, 0.1) and n = 0, 1, 2, ..., 20000.



FIGURE 2: Complex behaviors of the perturbed map *F* in the (x, y) space, where $\gamma = 0.0297$, the initial point is taken as (0.1, 0.1) and n = 0, 1, 2, ..., 20000.



FIGURE 3: Complex behaviors of the original map f in the (x, y, z) space, where the initial point is take as (0.1, 0.1, 0.1) and n = 0, 1, 2, ..., 20000.



FIGURE 4: Complex behaviors of the perturbed map *F* in the (x, y, z) space, where $\gamma = 0.0133$, the initial point is taken as (0.1, 0.1, 0.1) and n = 0, 1, 2, ..., 20000.

 Γ connecting the repellers z_1 and $z_2.$ Some constants that used to determine the range of perturbations are taken as follows:

$$\begin{split} \mu_{21} &= 1, \\ \delta_{10} &= 0.03 < \frac{r_1 - \|z_1 - x_{10}\|}{2} \approx 0.066987, \\ \delta_{20} &= 0.2 < \frac{r_2 - \|z_2 - x_{20}\|}{2} \approx 0.966987, \\ \delta_{11} &= 0.04 < \min\left\{\lambda_{10}\delta_{10}, \frac{r_2 - \|z_2 - x_{20}\|}{2} - \delta_{20}\right\} \\ &= 0.18, \\ \delta_{21} &= 0.2 < \lambda_{20}\delta_{20} \\ &= 0.8, \\ \delta_{22} &= 0.03 < \min\left\{\mu_{21}\delta_{21}, \frac{r_1 - \|z_1 - x_{10}\|}{2} - \delta_{10}\right\} \\ &= 0.036987, \\ \varepsilon_1 &= \frac{(\lambda_{10} - 1)\delta_{22}}{1 + \delta_{22}} \approx 0.145631, \\ \varepsilon_2 &= \frac{(\lambda_{20} - 1)\delta_{11}}{1 + \delta_{11}} \approx 0.115385, \\ \varepsilon_3 &= \min\left\{\varepsilon_1, \varepsilon_2, \frac{\delta_{10}\lambda_{10} - \delta_{11}}{1 + \delta_{10}}\right\} \\ &= 0.115385, \\ \varepsilon_4 &= \min\left\{\varepsilon_1, \varepsilon_2, \frac{\min\{\lambda_{20}\delta_{20} - \delta_{21}, \mu_{21}\delta_{21} - \delta_{22}\}}{1 + \max\{\delta_{20}, \delta_{21}\}}\right\} \\ &= 0.115385, \end{split}$$

 $\varepsilon_0 = \min\{\varepsilon_j, 1 \le j \le 4\} = 0.115385$. It is also easy to check that the perturbation *g* satisfies condition (10) for $|\gamma| \le 0.0133$. Then, it follows from the result of Theorem 5 that the perturbed system F = f + g also has a regular and nondegenerate heteroclinic cycle Γ' connecting repellers which is near to Γ . Consequently, *F* and *f* are chaotic in the sense of both Devaney and Li-Yorke.

The behaviors of the unperturbed map f with an initial point (x, y, z) = (0.1, 0.1, 0.1) are illustrated in Figure 3. We also do some computers simulations of F as γ increases from -0.0133 to 0 or from 0 to 0.0133, and find that all the simulations are also similar with that of the original map f in Figure 3. Here, we give one simulation of F with an initial point (x, y, z) = (0.1, 0.1, 0.1) for $\gamma = 0.0133$, see Figure 4. We can see that Figure 4 is also a small change to Figure 3, which shows that the heteroclinic cycle Γ' of F is near to Γ of

f. When we let $|\gamma|$ continuous to increase, we also find that the computer simulations gradually change until there is a big difference from that of the original map. This shows that the heteroclinic cycle Γ breaks or disappears.

Remark 9. In the above examples, it only needs the Lipschitz constant *L* and the values of *g* at z_i , x_{ij} for $1 \le i \le 2$, $0 \le j \le m_i - 1$ to satisfy condition (10), and does not need to compute the values of *g* at any other points. This is very easy to check out and is very convenient in applications. Since there are few literature giving concrete methods to identify an exact expanding area of a fixed point, it is very hard to get the largest perturbation range. But we think that the results obtained in this paper are also useful in practice. Because when a perturbation range ε_0 is determined as in the above examples, it can ensure that the persistence is maintained for a large range of parameters. The perturbation range obtained in these examples may not be the largest one for the persistence to be maintained. A more precise perturbed range is needed in practice and this will also be our further research.

5. Conclusions

(86)

In this paper, we studied persistence of heteroclinic cycles connecting repellers in Banach spaces. We proved that if a map with a regular and nondegenerate heteroclinic cycle connecting repellers undergoes a small Lipschitz perturbation, then the perturbed map still has a regular and nondegenerate heteroclinic cycle connecting repellers. Consequently, the perturbed map and the original map are simultaneously chaotic in the sense of both Devaney and Li-Yorke. We believe that the results obtained in the paper will be useful for studying the existence of chaos and will provide certain theoretical basis for practical applications of heteroclinic cycles of connecting repellers. Compared with some related papers, three major achievements on the persistence are summarized as follows. One is that the maps discussed in the paper only need to be continuous or continuously differentiable in some domains instead of the whole space. Since a lot of maps may not be continuous or continuously differentiable in the whole space, our results are more general in practice than those in some related papers. The second is that an explicit expression for the range of perturbations is given, while most related papers did not give such an expression. The expression is determined by some properties of the original maps. It only needs to check out some values of the perturbation map at certain points in practice. This is very convenient and has great potential in applications. The third is that different repellers are allowed to expand in different norms in \mathbf{R}^n , while some related papers only used the single Euclidean norm to do that. This is very meaningful since it is more general in practice for some fixed points to expand in different norms. To show the validity of the theoretical results, we give some illustrative examples. However, the range of perturbations obtained in this paper is only a sufficient condition for the persistence to be maintained, and it may not be the largest one. Since it is hard to determine the exact area of a fixed point and few researches have given concrete methods to do this, it is not easy to find the largest range of perturbations and this will be our further research.

Data Availability

The data in this paper are acquired from the corresponding author.

Conflicts of Interest

The author declares that there are no conflicts of interest.

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Research Article

Bifurcation Analysis and Exact Wave Solutions for the Double-Chain Model of DNA

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This work aims to study analytically the nonlinear model for deoxyribonucleic acid (DNA). Based on the complete discrimination and direct method, some new wave solutions are introduced. These solutions are sorted into solitary, periodic, kink (antikink), and singular solutions. Moreover, a part of them is illustrated graphically. Based on Hamilton concepts, we study the bifurcation and phase portrait for the Hamilton system corresponding to the model under consideration.

1. Introduction

Deoxyribonucleic acid (DNA) molecule carries the information for living beings that is required to live and propagate themselves. The nonlinear model for deoxyribonucleic acid (DNA) is attractive for the study because its properties can be investigated precisely by experiments gathering both the physical methods and biological tools [1]. In 1953, Watson and Crick [2] initially discovered the double helix construction of DNA; in spite of that, it is not easy until now to find a specific mathematical model that involved all its characteristics. The reason is its complicated structure and the existence of several motions such as the torsional, transverse, and longitudinal motions [3]. However, the existence of several motions of the DNA on thoroughly distinct time scales becomes aidable to model a few of these that predominate in the given time scale range. The idea that was introduced by Davydov [4] in his pioneer works related to the theoretical studies of the nonlinear characteristics of DNA has been firstly utilized by Englander and coauthors in 1980 to investigate dynamics of DNA regarding the nitrogen rational motion [5]. This idea has been further developed in several subsequent works. For instance, Yomosa proposed a

model of the dynamics plane of the base rotator [6], and this study was followed by Takeno and Homma who got better this model by considering the degree of freedom describing base rotations in the plane perpendicular to the helical axis around the structure of the backbone [7]. The denaturation process in which the base transverse motion along the hydrogen bond is regarded was investigated by Peyrard and Bishop [8]. Two kinds of internal motions, which have been proposed by Muto et al., contribute mainly to the denaturation process of DNA. These motions are longitudinal motions over the backbone and transverse motions over the hydrogen bond [9]. This model has been developed and improved in several works in order to study distinct motions and construct solitary wave-type solutions [10-14]. These waves acquire their significance from their ability to transmit energy without losing, i.e., the energy is conserved [8, 15], and moreover, they explicate the long-range interaction of kink solitons in the double chain [16, 17] and transcription regulation [18].

Taking into account some acceptable approximations from the point of biological science, the two equations describe a DNA model with double chains consisting of elastic two long homogeneous strands. These strands characterize two polynucleotide chains of DNA molecules attached by an elastic membrane which represents the hydrogen bonds between the base pair of two chains. The dynamical nonlinear system characterizing the double-chain model of DNA takes the form [19, 20]

$$I_{tt} - e_1^2 I_{xx} = a_1 I + b_1 I J + c_1 I^3 + d_1 I J^2,$$

$$J_{tt} - e_2^2 I_{xx} = a_2 J + b_2 I^2 + c_2 I^2 J + d_2 J^3 + m_0,$$
(1)

where *I* refers to the longitudinal displacement difference between the top and bottom wires, while *J* indicates the transverse displacements between the upper and lower strands, and e_i, a_i, b_i, c_i, d_i , and m_0 , i = 1, 2, are constants given by

$$e_{1} = \pm \frac{R_{1}}{\rho},$$

$$e_{2} = \pm \frac{R_{2}}{\rho},$$

$$a_{1} = \pm \frac{2\mu}{\rho\sigma h} (h - l_{0}),$$

$$a_{2} = \pm \frac{2\mu}{\rho\sigma},$$

$$b_{1} = 2b_{2} = \frac{2\sqrt{2}\mu l_{0}}{\rho\sigma h^{2}},$$

$$c_{1} = c_{2} = \frac{-2\mu l_{0}}{h^{3}\rho\sigma},$$

$$d_{1} = d_{2} = \frac{4\mu l_{0}}{h^{3}\rho\sigma},$$

$$m_{0} = \frac{\mu\sqrt{2}}{\rho\sigma} (h - l_{0}),$$
(2)

where ρ , σ , R_1 , and R_2 refer to density of mass, area of cross section, Young's modulus, and density of the tension of each strand, μ indicates the stiffness of the elastic membrane, h is the distance between the two strands, and l_0 is the height of membrane in the equilibrium.

To transform the nonlinear system (1) into a single partial differential equation, we present

$$J = \alpha I + \beta, \tag{3}$$

where α and β are two arbitrary constants, and furthermore, we assume $\beta = h/\sqrt{2}$ and $R_1 = R_2$. Thus, the linear system (1) is reduced to

$$I_{tt} - e_1^2 I_{xx} = f_3 I^3 + f_2 I^2 + f_1 I,$$
(4)

where f_i are arbitrary constants introduced for suitability, and they are given by

$$f_{3} = \frac{\omega_{0}}{h^{3}} (4\alpha^{2} - 2),$$

$$f_{2} = \frac{6\sqrt{2}\alpha\omega_{0}}{h^{2}},$$

$$f_{1} = \frac{6\omega_{0}}{h} - \frac{2\omega_{0}}{l_{0}},$$

$$\omega_{0} = \frac{l_{0}\mu}{\sigma\rho}.$$
(5)

The nonlinear model (1) has been investigated in several works. Riccati parameterized factorization method has been applied in [20] to construct some solitary wave solutions for the DNA model (1). The Φ^6 – model expansion method has been utilized in [21] to construct some solutions which are assorted into solitary, kink, and singular waves. Some solitary wave solutions for the double-chain model of DNA have been introduced and discussed [19]. Some exact wave solutions of this model have been constructed by using Conte's Painlevé truncation expansion and Pickering's truncation expansion [22]. Some bounded wave solutions for this model such as bell-shaped solitary waves and periodic waves have been formulated based on the method of the dynamical systems [23]. The generalized exponential rational function method has been applied to introduce exact form solutions and solitonic structures for this model [24].

Despite the wide variety of methods used to find wave solutions to nonlinear PDEs, the problem under study has a simple history, for instance, the bifurcation analysis [25–34], sine-Gordon expansion method [35, 36], Hirota bilinear technique [37], G'/G method [38–40], differential transform method (DTM), homotopy perturbation method (HPM) [41], Lie symmetry method [42], homotopic analysis method [43], trigonometric function series method [44], modified mapping method and extended mapping method [45], modified trigonometric function series method [46], tanh-coth expansion method and Jacobi function expansion method [47], and Jacobi elliptic function expansion method [48], and for more different techniques, see [49–53].

In this work, we are interested in constructing some traveling wave solutions for the nonlinear model (1) which is equivalent to building a wave solution for the reduced equation (1). We apply the complete discriminant system in addition to determining the intervals of permitted real propagations. The significance of finding these intervals enables us to construct only real wave solutions, and furthermore, for the same constraints on the system's parameters, there are several intervals of possible real wave propagations. Hence, the missing of such study in previous works leads to missing some wave solutions and the appearance of complex solutions. The bifurcation analysis is introduced which plays an important role in determining the types of the solutions before constructing them. We are also interested in studying the influence of the system's parameters on the solutions.

This work is organized as follows: Section 2 involves the reduction of the DNA model to ordinary differential equations and using the complete discrimination method to construct some traveling wave solutions for equation (1). Section 3 contains the study of some dynamical properties of equation (1) by employing the complete discrimination and Hamiltonian concepts. Section 4 is a graphic representation of some of the obtained solutions. Furthermore, it examines the influence of the one-parameter changing on the solutions keeping the other parameters fixed. Section 4 is a collection and the summary of the obtained results.

2. Exact Wave Solutions

Applying the wave transformation $I(x, t) = u(\zeta), \zeta = kx - \omega t$, to equation (4), we obtain

$$u'' = \frac{f_3}{\omega^2 - k^2 e_1^2} u^3 + \frac{f_2}{\omega^2 - k^2 e_1^2} u^2 + \frac{f_1}{\omega^2 - ka^2 e_1^2} u, \qquad (6)$$

where k is a constant that specifies the cosine of angle of propagation with ζ -axis and ω is an arbitrary constant that characterizes the speed of the wave and $\omega^2 - k^2 e_1^2 \neq 0$. For simplicity, we insert

$$u(\zeta) = p(\zeta) - \frac{f_2}{3f_3},$$
 (7)

into equation (6), and we get

$$p''(\zeta) = 2n_4 p^3(\zeta) + n_2 p(\zeta) + \frac{n_1}{2},$$
(8)

where ' indicates derivatives with respect to ζ and n_i , i = 2, 3, 4, are constants introduced for suitability and they are given by

$$n_{4} = \frac{f_{3}}{2(\omega^{2} - k^{2}e_{1}^{2})},$$

$$n_{2} = \frac{f_{2}^{2} - 3f_{3}f_{1}}{3f_{3}(k^{2}e_{1}^{2} - \omega^{2})},$$

$$n_{1} = \frac{2f_{3}(9f_{1}f_{2} - 2f_{2}^{2})}{27f_{3}^{2}(k^{2}e_{1}^{2} - \omega^{2})}.$$
(9)

Integrating both sides of equation (8) with respect to p, we obtain

$$p'^{2} = n_{4} (p^{4} + \gamma_{2} p^{2} + \gamma_{1} p + \gamma_{0}), \qquad (10)$$

where γ_i are arbitrary parameters. Separating the variables, we obtain the differential form

$$d\zeta = \frac{dp}{\sqrt{n_4 F_4(p)}},\tag{11}$$

where

$$F_4(p) = p^4 + \gamma_2 p^2 + \gamma_1 p + \gamma_0, \qquad (12)$$

in which

$$\gamma_2 = \frac{n_2}{n_4},$$
(13)
 $\gamma_1 = \frac{n_1}{n_4}.$

To integrate both sides of equation (11), the range of the parameters is required to be determined. The cause for this is that different values of the parameters imply different solutions to the integral. Hence, the key steps are to find the range of these parameters and consequently integrate both sides of equation (11). There are many tools utilized to find these ranges of parameters. In this work, we will apply a complete discrimination system for a polynomial. This method is a natural generalization of the discrimination $\Delta =$ $b^2 - 4ac$ for the quadratic polynomial $ax^2 + bx + c$, but it becomes difficult to calculate it for the higher degree polynomials. This problem had been solved with aid of computer algebra programs by Yang et al. by presenting an algorithm to compute the complete discrimination system for polynomial [54]. The complete discrimination system for the quartic polynomial $F_4(p) = p^4 + \gamma_2 p^2 + \gamma_1 p + \gamma_0$ is given in [55], and it admits the form

$$D_{1} = 4,$$

$$D_{2} = -\gamma_{2},$$

$$D_{3} = -2\gamma_{2}^{3} + 8\gamma_{2}\gamma_{0} - 9\gamma_{1}^{2},$$

$$E_{2} = 9\gamma_{2}^{2} - 32\gamma_{2}\gamma_{0},$$

$$D_{4} = -\gamma_{2}^{3}\gamma_{1}^{2} + 4\gamma_{0}\gamma_{2}^{4} + 36\gamma_{2}\gamma_{1}^{2} - 32\gamma_{2}^{2}\gamma_{0}^{2} - \frac{27}{4}\gamma_{1}^{4} + 64\gamma_{0}^{4}.$$
(14)

We study eight cases that describe different types of the roots for polynomial (12). To avoid confounding, we collect the classification of all the different types of the roots of the polynomial $F_4(p)$ by utilizing the discriminant system in Table 1. Furthermore, we integrate only on certain intervals for p in which $n_4F_4(p)$ is positive in order to get real solutions.

2.1. Case 1. Polynomial (12) has four real roots which are equal to zero if $D_2 = D_3 = D_4 = 0$. Hence, it is written as $F_4(p) = p^4$. Assuming $-\infty and <math>p(\zeta_0) = -\infty$ and integrating (11), we obtain

$$I(x,t) = -\frac{1}{\sqrt{n_4}(kx + \omega t - \zeta_0)} - \frac{\sqrt{2}ah}{2a^2 - 1},$$
 (15)

and consequently, we have

$$J(x,t) = \alpha \left[-\frac{1}{\sqrt{n_4} \left(kx + \omega t - \zeta_0 \right)} - \frac{\sqrt{2} ah}{2a^2 - 1} \right] + \beta.$$
(16)

Solutions (15) and (16) are singular solutions, and their singularity points lie on the plane $kx + ct - \zeta_0 = 0$.

TABLE 1: Types of the roots of the polynomial $F_4(p)$.

No.	Conditions on the discriminant system	Types of the roots for $F_4(p)$
1	$D_2 = D_3 = D_4 = 0$	All roots are equal to zero
2	$D_3 = D_4 = E_2 = 0$	Two real roots: one is triple and the other is simple
3	$D_3 = D_4 = 0, E_2 > 0$	Two double roots
4	$D_2 > 0, D_3 > 0, D_4 > 0$	Four real roots
5	$D_4 = 0, D_2 D_3 < 0$	One double root and two complex conjugate roots
6	$D_4 < 0, D_2 D_3 > 0$	Two real roots and two complex conjugate roots
7	$D_2 D_3 \le 0, D_4 > 0$	Two conjugate complex roots
8	$D_2 > 0, D_3 > 0, D_4 = 0$	Four real roots: one double and others simple

2.2. Case 2. If $D_3 = D_4 = E_2 = 0$, then the quartic polynomial (12) has two real roots: one is simple and the other is triple. Hence, it can be expressed as $F_4(p) = (p - p_1)^3 (p + 3p_1)$, where p_1 is assumed to be positive, i.e., $p_1 > 0$. We consider two subcases according to n_4 is positive or negative:

(i) For n₄ > 0, the possible interval for real propagation is p ∈] −∞, −3p₁[∪]p₁,∞[. Thus, if we choose p < −3p₁, assume p(ζ₀) = −∞, and integrate both sides of equation (11), we obtain

$$\sqrt{n_4} \int_{\zeta_0}^{\zeta} d\zeta = \int_{-\infty}^{p} \frac{dp}{(p-p_1)\sqrt{(p-p_1)(p+3p_1)}}.$$
 (17)

It follows

$$p(\zeta) = p_1 - \frac{p_1}{p_1 \sqrt{n_4}(\zeta - \zeta_0) + 1} + \frac{1}{\sqrt{n_4}(\zeta - \zeta_0)}.$$
 (18)

Thus, the solution of equation (1) becomes

$$I(x,t) = p_1 - \frac{p_1}{p_1\sqrt{n_4}(\zeta - \zeta_0) + 1} + \frac{1}{\sqrt{n_4}(\zeta - \zeta_0)} - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[p_1 - \frac{p_1}{p_1\sqrt{n_4}(\zeta - \zeta_0) + 1} + \frac{1}{\sqrt{n_4}(\zeta - \zeta_0)} - \frac{f_2}{3f_3} \right] + \beta.$$
(19)

Similarly, we can calculate the solution if $p \in]p_1, \infty[$.

(ii) If $n_4 < 0$, the possible interval for p to obtain real propagation is $p \in]-3p_1, p_1[$. Thus, we assume $p(\zeta_0) = -3p_1$ and integrate both sides of equation (11), and we get

$$I(x,t) = p_1 - \frac{4p_1}{1 + 4n_4 p_1^2 (\zeta - \zeta_0)^2} - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[p_1 - \frac{4p_1}{1 + 4n_4 p_1^2 (\zeta - \zeta_0)^2} - \frac{f_2}{3f_3} \right] + \beta.$$
(20)

Both solutions (19) and (20) are singular solutions for equation (1).

2.3. *Case 3*. The polynomial $F_4(p)$ has two double real zeros, namely, $\pm p_1$, where $p_1 > 0$, if $D_3 = D_4 = 0$, $E_2 > 0$, $D_2 > 0$. Hence, it can be introduced as $F_4(p) = (p^2 - p_1^2)^2$. We consider the following two cases in which n_4 is either positive or negative.

(i) If $n_4 > 0$, the intervals for real propagation are $p < -p_1$, $-p_1 , and <math>p > p_1$. If we consider the case in which $p < -p_1$ and assume $p(\zeta_0) = -\infty$, equation (11) gives

$$\sqrt{n_4} \int_{\zeta_0}^{\zeta} d\zeta = \int_{-\infty}^{p} \frac{\mathrm{d}p}{p^2 - p_1^2}.$$
 (21)

It follows

$$p = -p_1 \operatorname{coth} \left(p_1 \sqrt{n_4} \left(\zeta - \zeta_0 \right) \right). \tag{22}$$

Using equations (7) and (3), we obtain a wave solution for equation (1) in the form

$$I(x,t) = -p_1 \coth(p_1 \sqrt{n_4} (\zeta - \zeta_0)) - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[-p_1 \coth(p_1 \sqrt{n_4} (\zeta - \zeta_0)) - \frac{f_2}{3f_3} \right] + \beta.$$
(23)

When $p > p_1$, equation (1) has the same solution shown in equation (23) if $p_1 \rightarrow -p_1$. Similarly, if we choose $p \in]-p_1, p_1[$ and assume $p(\zeta_0) = 0$, equation (1) has a solution in the form

$$I(x,t) = -p_{1} \tanh(p_{1}\sqrt{n_{4}}(\zeta - \zeta_{0})) - \frac{f_{2}}{3f_{3}},$$

$$J(x,t) = \alpha \left[-p_{1} \tanh(p_{1}\sqrt{n_{4}}(\zeta - \zeta_{0})) - \frac{f_{2}}{3f_{3}}\right] + \beta.$$
(24)

(ii) The case in which $n_4 < 0$ is excluded since $n_4F_4(p) < 0$ for all $p \in \mathbb{R}$.

2.4. *Case* 4. The polynomial $F_4(p)$ has four real zeros, namely, $p_1, p_2, p_3, -(p_1 + p_2 + p_3)$, where we assumed $0 < p_1 < p_2 < p_3$ if $D_2 > 0$, $D_3 > 0$, $D_4 > 0$. Therefore, it takes the form $F_4(p) = (p - p_1)(p - p_2)$ $(p - p_3)(p + p_3)$

 $p_1 + p_2 + p_3$). Now, we consider the two cases $n_4 > 0$ and $n_4 > 0$, individually:

(i) When $n_4 > 0$, the possible intervals of p for real propagation are $p < -(p_1 + p_2 + p_3)$, $p_1 , <math>p_1 , <math>p_2 < -(p_1 + p_2 + p_3)$, $p_2 < -(p_1 + p_2 + p_3)$, $p_2 < -(p_1 + p_3)$, $p_3 < -(p_1 + p_3)$.

 p_2 , and $p > p_3$. If we choose $p < -(p_1 + p_2 + p_3)$ and assume $p(\zeta_0) = -(p_1 + p_2 + p_3)$, equation (11) becomes

$$\sqrt{n_4} \int_{\zeta_0}^{\zeta} d\zeta = \int_{-(p_1+p_2+p_3)}^{p} \frac{\mathrm{d}p}{\sqrt{(p-p_1)(p-p_2)(p-p_3)(p+p_1+p_2+p_3)}}.$$
(25)

It implies to

$$p = p_1 - \frac{(p_1 - p_3)(2p_1 + p_2 + p_3)}{p_1 - p_3 + (p_1 + p_2 + 2p_3)\operatorname{sn}^2(\Omega_1(\zeta - \zeta_0), k_1)}, \quad \zeta_0 < \zeta < \zeta_1,$$
(26)

where $\begin{aligned} &\Omega_1 = 1/2\sqrt{n_4(p_3 - p_1)(p_1 + 2p_2 + p_3)}, \\ &k_1 = \sqrt{((p_2 - p_1)(p_1 + p_2 + 2p_3))} \quad ((p_3 - p_1)(p_1 + 2p_2 + p_3)), \\ &+ 2p_2 + p_3)), \text{ and } \zeta_1 = (1/\Omega_1)K(k_1). \quad K(k_1) \text{ is a} \end{aligned}$

complete elliptic integral of the first type [56]. Using equations (7) and (3), we obtain a new traveling wave solution for equation (1) in the form

$$\Omega_{1}(x,t) = p_{1} - \frac{(p_{1} - p_{3})(2p_{1} + p_{2} + p_{3})}{p_{1} - p_{3} + (p_{1} + p_{2} + 2p_{3})\mathrm{sn}^{2}(\Omega_{1}(\zeta - \zeta_{0}), k_{1})} - \frac{f_{2}}{3f_{3}},$$

$$J(x,t) = \alpha \left[p_{1} - \frac{(p_{1} - p_{3})(2p_{1} + p_{2} + p_{3})}{p_{1} - p_{3} + (p_{1} + p_{2} + 2p_{3})\mathrm{sn}^{2}(\Omega_{1}(\zeta - \zeta_{0}), k_{1})} - \frac{f_{2}}{3f_{3}} \right] + \beta.$$
(27)

If we select $p_1 , postulate <math>p(\zeta_0) = p_1$, and follow the same procedures, we will obtain a new traveling wave solution for equation (1) in the form

$$I(x,t) = -p_1 - p_2 - 2p_3 + \frac{(p_1 + 2p_2 + p_3)(2p_1 + p_2 + 2p_3)}{p_1 + 2p_2 + p_3 + (p_1 - p_2)\operatorname{sn}^2(\Omega_1(\zeta - \zeta_0), k_1)} - \frac{f_2}{3f_3},$$

$$J(x,t)\alpha \left[-p_1 - p_2 - 2p_3 + \frac{(p_1 + 2p_2 + p_3)(2p_1 + p_2 + 2p_3)}{p_1 + 2p_2 + p_3 + (p_1 - p_2)\operatorname{sn}^2(\Omega_1(\zeta - \zeta_0), k_1)} - \frac{f_2}{3f_3} \right] + \beta,$$
(28)

where $\zeta_0 < \zeta < \zeta_1$. Also, if we elect $p > p_3$ and suppose $p(\zeta_0) = p_3$, we will get a new wave solution for equation (1) in the form

$$I(x,t) = p_{2} + \frac{(p_{2} - p_{3})(p_{1} + 2p_{2} + p_{3})}{(p_{1} + p_{2} + 2p_{3})\mathrm{sn}^{2}(\Omega_{1}(\zeta - \zeta_{0}), k_{1})} - \frac{f_{2}}{3f_{3}},$$

$$J(x,t) = \alpha \left[p_{2} + \frac{(p_{2} - p_{3})(p_{1} + 2p_{2} + p_{3})}{(p_{1} + p_{2} + 2p_{3})\mathrm{sn}^{2}(\Omega_{1}(\zeta - \zeta_{0}), k_{1})} - \frac{f_{2}}{3f_{3}} \right] + \beta,$$
(29)

where $\zeta_0 < \zeta < \zeta_1$.

(ii) If $n_4 < 0$, the allowed intervals of p for real propagation are $-p_1 - p_2 - p_3 or <math>p_2 .$ Thus, if we chose $p_1 and assume <math>p(\zeta_0) = -p_1 - p_2 - p_3$, equation (11) takes the form

$$\sqrt{-n_4} \int_{\zeta_0}^{\zeta} d\zeta = \int_{-p_1 - p_2 - p_3}^{p} \frac{\mathrm{d}p}{\sqrt{\sqrt{-(p - p_1)(p_2 - p)(p_3 - p)(p + p_1 + p_2 + p_3)}}}.$$
(30)

It gives

$$p = p_3 + \frac{(p_1 - p_3)(p_1 + p_2 + 2p_3)}{p_3 - p_1 + (2p_1 + p_2 + p_3)\operatorname{sn}^2(\Omega_2(\zeta - \zeta_0), k_2)}, \quad \zeta_0 < \zeta < \zeta_2,$$
(31)

where $\Omega_2 = \frac{1}{2}\sqrt{-n_4(p_3 - p_1)(2p_2 + p_1 + p_3)}, k_2 = \sqrt{((p_3 - p_2)(2p_1 + p_2 + p_3))/((p_3 - p_1)(2p_2 + p_1))}$

+ p_3)), and $\zeta_2 = (1/\Omega_2)K(k_2)$. Utilizing equations (7) and (3), we obtain a new solution for equation (1):

$$I(x,t) = p_{3} + \frac{(p_{1} - p_{3})(p_{1} + p_{2} + 2p_{3})}{p_{3} - p_{1} + (2p_{1} + p_{2} + p_{3})\operatorname{sn}^{2}(\Omega_{2}(\zeta - \zeta_{0}), k_{2})} - \frac{f_{2}}{3f_{3}},$$

$$J(x,t) = \alpha \left[p_{3} + \frac{(p_{1} - p_{3})(p_{1} + p_{2} + 2p_{3})}{p_{3} - p_{1} + (2p_{1} + p_{2} + p_{3})\operatorname{sn}^{2}(\Omega_{2}(\zeta - \zeta_{0}), k_{2})} \right] + \beta.$$
(32)

Similarly, if we select $p_2 and assume <math>p(\zeta_0) = p_2$, we present a new wave solution for equation (1) in the form

$$I(x,t) = p_1 - \frac{(p_1 - p_2)(p_1 - p_3)}{p_1 - p_3 + (p_3 - p_2)\operatorname{sn}^2(\Omega_2(\zeta - \zeta_0), k_2)} - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[p_1 - \frac{(p_1 - p_2)(p_1 - p_3)}{p_1 - p_3 + (p_3 - p_2)\operatorname{sn}^2(\Omega_2(\zeta - \zeta_0), k_2)} - \frac{f_2}{3f_3} \right] + \beta.$$
(33)

2.5. *Case* 5. The polynomial $F_4(p)$ has one double real root and two conjugate complex roots if $D_4 = 0$ and $D_2D_3 < 0$. Therefore, it takes the form $F_4(p) = (p - p_1)^2 (p - p_2)(p - p_2^*)$, where * refers to the complex conjugate and $p_1 = -\text{Re}p_2$. We consider the case in which $n_4 > 0$, and sequentially, the allowed intervals for real propagation are $p < p_1$ or $p > p_1$. Choosing $p > p_1$, assuming $p(\zeta_0) = \infty$, and integrating both sides of equation (11), we obtain

$$p = p_1 + \frac{4p_1^2 + \rho^2}{-2p_1 + \rho \sinh\left(\sqrt{n_4(4p_1^2 + \rho^2)} \ (\zeta - \epsilon)\right)},$$
(34)

where $\epsilon = \zeta_0 - s (inh^{-1} (2p_1/\rho))/\sqrt{n_4 (4p_1^2 + \rho^2)}$ is a new constant which is introduced for suitability and $\rho = Imp_2$. Employing equations (7) and (3), we obtain a solution for equation (1) in the form

$$I(x,t) = p_{1} + \frac{4p_{1}^{2} + \rho^{2}}{-2p_{1} + \rho\sinh\left(\sqrt{n_{4}(4p_{1}^{2} + \rho^{2})} (\zeta - \epsilon) - \frac{f_{2}}{3f_{3}}\right)},$$

$$J(x,t) = \alpha \left[p_{1} + \frac{4p_{1}^{2} + \rho^{2}}{-2p_{1} + \rho\sinh\left(\sqrt{n_{4}(4p_{1}^{2} + \rho^{2})} (\zeta - \epsilon) - \frac{f_{2}}{3f_{3}}\right) + \beta.$$
(35)

It can be noted that the case in which n_4 is negative does not work because $F_4(p) \ge 0$ for all $p \in \mathbb{R}$.

2.6. *Case* 6. The polynomial $F_4(P)$ has two real roots and two complex conjugate roots if $D_4 < 0$ and $D_2D_3 > 0$. Hence, it can be written in the form $F_4(p) = (p - p_1)(p - p_2)(p - p_3)(p - p_3^*)$, where $p_1 < p_2$ and $\operatorname{Re} p_3 = (1/2)(p_1 + p_2)$. We consider the following:

(i) If $n_4 > 0$, then the permitted intervals for real propagation are $p > p_2$ and $p < p_1$. Selecting $p < p_2$

and $p(\zeta_0) = p_2$ and integrating both sides of equation (11), we obtain

$$p = \frac{p_2 B_1 - p_1 A_1 + (p_2 B_1 + p_1 A_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} (\zeta - \zeta_0), k_2\right)}{(A_1 + B_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} (\zeta - \zeta_0), k_2\right) - (A_1 - B_1)},$$
(36)

where $k_2 = \sqrt{((A_1 + B_1)^2 - (p_2 - p)_1^2)/4A_1B_1}$ and $A_1^2 = B_1^2 = (1/4)(p_1 - p_2)^2 + \text{Im}^2(p_3)$. Taking into account equations (7) and (3), we obtain a novel wave solution for equation (1) in the form

$$I(x,t) = \frac{p_2 B_1 - p_1 A_1 + (p_2 B_1 + p_1 A_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} \left(kx + \omega t - \zeta_0\right), k_2\right)}{(A_1 + B_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} \left(\zeta - \zeta_0\right), k_2\right) - (A_1 - B_1)} - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[\frac{p_2 B_1 - p_1 A_1 + (p_2 B_1 + p_1 A_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} \left(kx + \omega t\right), k_2\right)}{(A_1 + B_1) \operatorname{cn} \left(\sqrt{n_4 A_1 B_1} \left(\zeta - \zeta_0\right), k_2\right) - (A_1 - B_1)} - \frac{f_2}{3f_3}\right] + \beta.$$
(37)

(ii) If $n_4 < 0$, then the allowed intervals of possible propagation are $p_1 . Assuming <math>p(\zeta_0) = p_1$

and integrating both sides of equation (11), we obtain

$$p = \frac{p_2 B_2 + p_1 A_2 + (p_1 A_2 - p_2 B_2) \operatorname{cn} \left(\sqrt{-n_4 A_2 B_2} \left(\zeta - \zeta_0\right), k_3\right)}{B_2 + A_2 - (B_2 - A_2) \operatorname{cn} \left(\sqrt{-n_4 A_2 B_2} \left(\zeta - \zeta_0\right), k_3\right)},$$
(38)

where $k_3 = \sqrt{((p_2 - p_1)^2 - (A_2 - B_2)^2)/4A_2B_2}$, $A_2^2 = (1/4)(p_1 + 3p_2)^2 + \text{Im}^2 p_3$, and $B_2^2 = (1/4)(3p_1 + p_2)^2 + \text{Im}^2 p_3$. Utilizing equations

(7) and (3), we construct a novel wave solution for equation (1) in the form

$$I(x,t) = \frac{p_2 B_2 + p_1 A_2 + (p_1 A_2 - p_2 B_2) \operatorname{cn} (\sqrt{-n_4 A_2 B_2} (kx + \omega t - \zeta_0), k_3)}{B_2 + A_2 - (B_2 - A_2) \operatorname{cn} (\sqrt{-n_4 A_2 B_2} (kx + \omega t - \zeta_0), k_3)} - \frac{f_2}{3f_3},$$

$$J(x,t) = \alpha \left[\frac{p_2 B_2 + p_1 A_2 + (p_1 A_2 - p_2 B_2) \operatorname{cn} (\sqrt{-n_4 A_2 B_2} (kx + \omega t - \zeta_0), k_3)}{B_2 + A_2 - (B_2 - A_2) \operatorname{cn} (\sqrt{-n_4 A_2 B_2} (kx + \omega t - \zeta_0), k_3)} - \frac{f_2}{3f_3} \right] + \beta.$$
(39)

2.7. *Case* 7. The polynomial $F_4(p)$ has two conjugate complex roots, namely, p_1, p_1^*, p_2, p_2^* , if $D_2D_3 \le 0$ and $D_4 > 0$. Therefore, it is expressed as $F_4(p) = (p - p_1)(p - p_1^*)(p - p_2)(p - p_2^*)$, where

Re. $p_1 = -\text{Re. } p_2$. The permitted interval for real propagation for the case $n_4 > 0$ is $p \in \mathbb{R}$. Thus, we follow similar steps as above and obtain a new traveling wave solution for equation (1) as

$$I(x,t) = \frac{\operatorname{Re}p_{1} + \operatorname{Im}p_{1}\delta + (\operatorname{Im}p_{1} + \operatorname{Re}p_{1}\delta)\operatorname{tn}((\sqrt{n_{4}}/2)(A_{2} + B_{2})(kx + \omega t - \zeta_{0}), k_{4})}{\delta + \operatorname{tn}((\sqrt{n_{4}}/2)(A_{2} + B_{2})(kx + \omega t - \zeta_{0}), k_{4})} - \frac{f_{2}}{3f_{3}},$$

$$J(x,t) = \alpha \left[\frac{\operatorname{Re}p_{1} + \operatorname{Im}p_{1}\delta + (\operatorname{Im}p_{1} + \operatorname{Re}p_{1}\delta)\operatorname{tn}((\sqrt{n_{4}}/2)(A_{2} + B_{2})(kx + \omega t - \zeta_{0}), k_{4})}{\delta + \operatorname{tn}((\sqrt{n_{4}}/2)(A_{2} + B_{2})(kx + \omega t - \zeta_{0}), k_{4})} - \frac{f_{2}}{3f_{3}} \right] + \beta,$$
(40)

where $A_2^2 = [\text{Im}p_1 - \text{Im}p_2]^2$, $B_2^2 = [\text{Im}p_1 - \text{Im}p_2]^2 + 4\text{Re}^2 p_1$, $\delta^2 = (4\Re p_1^2 - (A_2 - B_2)^2)/((A_2 + B_2)^2 - 4\Re p_1^2)$, and $k_4 = (2\sqrt{A_2B_2})/A_2 + B_2$. It can be noted that the case in which $n_4 < 0$ does not work because $F_4(p) > 0$ for all $p \in \mathbb{R}$.

2.8. *Case 8.* The polynomial $F_4(p)$ has four real roots in which one of them is double and the others are simple if $D_2 > 0, D_3 > 0$, and $D_4 = 0$. Hence, it takes the form $F_4(p) = (p - p_1)^2 (p - p_2) (p - p_3)$, where $p_1 < p_2 < p_3$ and

 $p_3 = -(2p_1 + p_2)$. We consider the two cases in which n_4 is either positive or negative:

(i) If $n_4 > 0$, then the possible intervals for real propagation are $p < p_1, p > p_3$, and $p_1 . With similar computations as in previous cases, we present the solution of equation (1) directly.$

If we choose $p > -(2p_1 + p_2)$ and assume $p(\zeta_0) = -(2p_1 + p_2)$, we have a new wave solution for equation (1) in the form

$$I(x,t) = -2p_1 - p_2 + 4(p_2 - p_1)\operatorname{sech}\left(\sqrt{n_4(-3p_1 - p_2)}(kx + \omega t - \zeta_0) - \frac{f_2}{3f_3}, (41)\right)$$

$$J(x,t) = \alpha \left[-2p_1 - p_2 + 4(p_2 - p_1)\operatorname{sech}\left(\sqrt{n_4(-3p_1 - p_2)}(kx + \omega t - \zeta_0) - \frac{f_2}{3f_3}\right] + \beta.$$

In similar calculations, we can calculate the wave solution for $p < p_1$ and $p_1 .$

(ii) If $n_4 < 0$, the allowed interval for real propagation is $p \in]p_2, p_3[$, and postulating $p(\zeta_0) = p_2$, we obtain a new wave solution for equation (1) in the form

$$I(x,t) = p_{1} + \frac{(p_{2} - p_{1})(3p_{1} + p_{2})}{2p_{1} + (p_{1} + p_{2})\cosh\sqrt{-n_{4}(3p_{1} + p_{2})(p_{1} - p_{2})}(kx - \omega t - \zeta_{0})} - \frac{f_{2}}{3f_{3}},$$

$$I(x,t) = \alpha \left[p_{1} + \frac{(p_{2} - p_{1})(3p_{1} + p_{2})}{2p_{1} + (p_{1} + p_{2})\cosh\sqrt{-n_{4}(3p_{1} + p_{2})(p_{1} - p_{2})}(kx - \omega t - \zeta_{0})} - \frac{f_{2}}{3f_{3}} \right] + \beta.$$

$$V(q) = -\frac{n_{4}}{2} \left(p^{4} + \gamma_{2}p^{2} + \gamma_{1}p \right),$$
(42)
(42)
(42)

3. Dynamic Properties

The aim of this section is to investigate some dynamic properties for equation (4) by investigating the bifurcation and phase portrait for the traveling wave system corresponding to equation (8) which takes the form

$$p' = z,$$

$$z' = 2n_4 \left(p^3 + \frac{\gamma_2}{2} p + \frac{\gamma_1}{4} \right).$$
(43)

System (43) is a Hamiltonian system with one degree of freedom related to Hamilton function:

$$\frac{1}{2}z^2 + V(p) = h, (44)$$

where h is an arbitrary constant and

is the potential function. It is well known that the equilibrium points for the Hamilton system (43) are also critical points for the potential function (45), i.e., they are the roots of

$$\frac{dV}{dp} = -2n_4 \left(p^3 + \frac{\gamma_2}{2} p + \frac{\gamma_1}{4} \right).$$
(46)

Thus, we use the discrimination of (46) to determine the number of equilibrium points. The discrimination of (46) is

$$\Delta = -\frac{1}{8} \left[\frac{\gamma_1^2}{8} + \frac{\gamma_2^3}{27} \right]. \tag{47}$$

Now, let us determine the number of equilibrium points for the Hamilton system (43) and study the properties of its phase space. Thus, we need to define energy curve corresponding to

$$\mathscr{C}_{h} = \{(p, z) \in \mathbb{R}^{2} : z^{2} = 2(h - V(p))\}.$$
 (48)

It is well known that any orbit for the Hamilton system (43) is an energy curve on a certain level of the energy.

Case 1. The dynamical system (43) has a unique equilibrium point if dV/dp = 0 has a unique real root. This happens in two cases which are studied individually:

- (i) If Δ = 0 and γ₂ = 0, then dV/dp = 0 has one triple real root, i.e., dV/dp = -2n₄p³. This shows (0,0) is a unique equilibrium point for system (43) which is saddle if n₄ > 0, and it is center if n₄ < 0. The phase space for this case is outlined by Figures 1(a) and 1(b). The value of the energy at the equilibrium point (0,0) is h₁ = V (0,0) = 0. The following proposition describes Figure 1.
- (ii) If $\Delta < 0$, then dV/dp = 0 has one real zero and two complex conjugate roots, i.e.,

$$\frac{\mathrm{d}V}{\mathrm{d}p} = -2n_4\left(p-a\right)\left[\left(p+\frac{a}{2}\right)^2 + m^2\right].$$
(49)

It is clear that the point (a, 0) is a unique equilibrium point for the Hamilton system (43) and it is saddle if $n_4 > 0$ and center if $n_4 < 0$. The phase space is clarified by Figures 2(a) and 2(b).

Proposition 1. The Hamiltonian system (43) has a unique equilibrium point (0,0) if $\gamma_1 = \gamma_2 = 0$. If $n_4 > 0$, it is a saddle point and all the orbits are unbounded, see Figure 1(a). If $n_4 < 0$, system (43) has a family of bounded periodic orbits $\{\mathscr{C}_h: h > h_1\}$ about the center point (0,0) as outlined by Figure 1(b). A similar conclusion can be presented to describe Figure 2.

Case 2. If $\Delta = 0$ and $\gamma_2 < 0$, then dV/dp = 0 has two real roots: one is simple and the other is double. Thus, we write $dV/dp = -2n_4(p-a)^2(p+2a)$ and so (a,0) and (-2a,0) are two equilibrium points for system (43). It is clear that (a,0) is a cusp while (-2a,0) is a center if $n_4 < 0$ and saddle if $n_4 > 0$. The phase portrait for this case is outlined in Figure 3. The following proposition gives a short description for the phase portrait for this case.

Proposition 2. The Hamiltonian system (43) has two equilibrium points $E_1 = (a, 0)$ and $E_2(-2a, 0)$. If $n_4 > 0$, then E_1 is cusp point while E_2 is saddle, and furthermore, all the phase space orbits are unbounded as outlined in Figure 3(a). While if $n_4 < 0$, E1 is a cusp and E_2 is a center. The Hamilton system (43) has two bounded families of periodic orbits which are illustrated in green and blue and separated by the phase curve $\{\mathscr{C}_h: h = V(a, 0)\}$ in red.

Case 3. If $\Delta > 0$ and $\gamma_2 < 0$, then dV/dp has three real roots, i.e., it can be written as $dV/dp = -2n_4(p-a)(p-b)$ (p+a+b), where we assumed b > a > 0. Consequently, the dynamical system (43) has three equilibrium points (a, 0), (b, 0), and (-a - b, 0). If $n_4 > 0$, then (a, 0) is center and (b, 0) and (-a - b, 0) are saddle points. While if $n_4 < 0$,

Proposition 3. If $\Delta > 0$ and $\gamma_2 < 0$, then the Hamilton system has three equilibrium points (a, 0), (b, 0), (43) and (-a - b, 0). If $n_4 > 0$, there are two families of orbits $\{\mathscr{C}_h: h \in [V(a), V(b)]\}$ in green in which one of them is bounded and surrounded by the homoclinic orbit $\{\mathscr{C}_h: h = V(2a)\}$ while the other family is unbounded. Moreover, all the other orbits are unbounded, see Figure 4(a)for more clarification. If $n_4 < 0$, system (43) has three equilibrium points in which one is a saddle and the others are centers. It has three bounded families of periodic orbits. Two of them in blue and green are periodic orbits around the two center points (2a, 0) and (-a - b, 0), and they are separated by the homoclinic orbit in red $\{\mathscr{C}_h: h = V(a)\}$. The third one is a family of superperiodic orbits in brown $\{\mathscr{C}_h: h > V(a)\}$ around the two centers points and lies outside the homoclinic orbit in red. For more details about superperiodic orbits, see, for example, [57].

The investigation of the type of the phase space orbits is helpful in determining the types of the solutions. For instance, the existence of periodic orbits, homoclinic orbits, and heteroclinic orbits for the traveling wave system (43) indicates the existence of periodic wave solutions, solitary, and kink solutions for equation (1). Furthermore, this analysis can be employed to construct the traveling wave solution by introducing the constraints on the coefficients of function (12). Consequently, we can prove the following theorem.

Theorem 1. Let $I(x,t) = u(kx - \omega t)$ and $J(x,t) = \alpha I(x,t) + \beta$ be a solution for the double-chain model of DNA (1), then

- (i) It is a periodic solution if (a) $n_4 < 0, h > 0, \Delta = 0 (\gamma_1 = \gamma_2 = 0),$ (b) $n_4 < 0, h > V (a, 0), \Delta < 0,$ (c) $n_4 < 0,$ $\Delta = 0, \gamma_2 < 0, h \in]V (-2a), V(a)[\cup]V(a), \infty[,$ (d) $n_4 > 0, \Delta > 0, \gamma_2 > 0, h \in]V(a), V(b)[,$ (e) $n_4 < 0, \Delta > 0, \gamma_2 > 0, h \in]V(a), \infty[,$ (f) $n_4 < 0, \Delta > 0, \gamma_2 > 0, h \in]V(a), \infty[,$ [V(a), $\infty[\cup]V(b), V(a)[\cup]V(c), V(b)[.$
- (ii) It is a solitary wave solution if $n_4 < 0, \Delta > 0$, $\gamma_2 < 0$, and h = V(a).
- (iii) It is a kink (antikink) solution if $n_4 > 0, \Delta > 0, \gamma_2 < 0, and h = V(b)$.

4. Graphic Interpretations

This section aims to illustrate some of the obtained solutions graphically. Moreover, we study the influence of the physical parameters on the obtained solutions by considering two



FIGURE 1: Phase portrait for the Hamilton system (43) for $\gamma_1 = 0$ and $\gamma_2 = 0$. (a) $n_4 = 1$ and (b) $n_4 = -1$.



FIGURE 2: Phase portrait for the Hamilton system (43) for $\Delta < 0$. (a) $n_4 = 1$, $\gamma 1 = -16$, and $\gamma 2 = -4$. (b) $n_4 = -1$, $\gamma_1 = -16$, and $\gamma_2 = -4$.

types of solutions: one of them is kink solution and the other is periodic.

Figure 5 and 6 illustrates the kink solution (24) for different aspects. Figures 5(a) and 5(b) clarify the 3D and contour representation for the kink solution (24) when k = 1, $\omega = 2$, $\sigma = 0.001$, $\rho = 0.1$, $\mu = 0.0001$, h = 0.002, R_1 = 0.1, and $l_0 = 0.002$. Now, we illustrate graphically the influence of some parameters on the kink solution while the other parameters are fixed. Figure 6(a) illustrates the effect of the change of the distance between the two strands. It is remarkable the amplitude of solution (24) is decreased when the distance between the two strands is increased. Figure 6(b) clarifies the amplitude of the solution is increased when the stiffness of the elastic membrane is increased. Figure 6(c) clarifies the amplitude of the solution is decreased when the area of the cross section of each strand is increased. Figure 6(d) outlines the amplitude of the kink solution (24) is increased when the height of the membrane in the equilibrium is increased (see Figure 7).

Now, we are going to clarify solution (28) graphically and study the influence of parameter changes on solution (28). Solution (28) is periodic as outlined in Figure 8(a), and its contour is illustrated in Figure 8(b). If the distance between the two strands is increased and the other parameters are fixed, the amplitude of the solution is unchanged but the width of the solution is increased as outlined in Figure 8(a). The amplitude of solution (28) is not affected by the changes in the stiffness of the elastic membrane, while the width of the solution is decreased when the stiffness of the elastic membrane is increased as clarified by Figure 8(d). If the area of the cross section of each strand is increased, then the



FIGURE 3: Phase portrait for the Hamilton system (43) for $\Delta = 0$ and $\gamma_2 < 0$. (a) $n_4 = 1$, $\gamma_1 = 8$, and $\gamma_2 = -6$. (b) $n_4 = -1$, $\gamma_1 = 8$, and $\gamma_2 = -6$.



FIGURE 4: Phase portrait for the Hamilton system (43) for $\Delta > 0$ and $\gamma_2 < 0$. (a) $n_4 = 1$, $\gamma_1 = 8$, and $\gamma_2 = -6$. (b) $n_4 = -1$, $\gamma_1 = 8$, and $\gamma_2 = -6$.



FIGURE 5: Graphic representation of solution (24) for $R_1 = 0.1$, h = 0.002, k = 1, $l_0 = 0.002$, $\mu = 0.0001$, $\omega = 2$, $\rho = 0.1$, and $\sigma = 0.001$. (a) 3D graphic and (b) 2D contour plot.



FIGURE 6: Parameters affecting solution (24): (a) changes in h for $R_1 = 0.1$, k = 1, $l_0 = 0.001$, $\mu = 0.002$, $\rho = 0.1$, and $\sigma = 0.0001$, (b) changes in μ for $R_1 = 0.1, h = 0.002, k = 1, l_0 = 0.001, \omega = 2, \rho = 0.1, \text{ and } \sigma = 0.0001,$ (c) changes in for σ $R_1 = 0.1, h = 0.002, k = 1, l_0 = 0.001, \mu = 0.0001, \omega = 0.1, \text{ and } \rho = 0.1,$ for and (d) changes in l_0 $R_1 = 0.1, h = 0.002, k = 1, \sigma = 0.001, \rho = 0.1, \mu = 0.0001, \text{ and } \omega = 2.$

amplitude keeps fixed while the width is increased, see Figure 8(d). Figure 9(b) outlines the amplitude of the periodic solution (28) is kept unchanged while its width is decreased when the height of the membrane in the equilibrium is increased. Figure 9(a) illustrates the influence of the superperiodic wave solution (39) due the changes in the stiffness of the elastic membrane. If the stiffness of the elastic membrane increases, the amplitude is kept unchanged while the width decreases. Figure 9(b) clarifies the influence of distance between the two strands on the superwave solution (39). If distance between the two strands is increased, then the amplitude and the width of superwave solution (39) are increased.

It is worth mentioning that we can make the same study for the remaining obtained solutions. However, we only give the 3D graphic and the 2D counter for some solutions. The 3D graphic representation and the singularity plane are outlined in Figure 10(a), while Figure 10(b) illustrates the 2D contour of solution (15). Solution (37) is illustrated in Figure 11.



FIGURE 7: Graphic representation of solution (28) for $R_1 = 0.1$, h = 0.002, k = 1, $l_0 = 0.002$, $\mu = 0.0001$, $\omega = 0.2$, $\rho = 0.1$, and $\sigma = 0.001$. (a) 3D graphic and (b) 2D contour plot.



FIGURE 8: Continued.



FIGURE 8: Parameters affecting solution (28): (a) changes in *h* for $R_1 = 0.1$, k = 1, $l_0 = 0.001$, $\mu = 0.002$, $\omega = 2$, $\rho = 0.1$, and $\sigma = 0.0001$, (b) changes in μ for $R_1 = 0.1$, h = 0.002, k = 1, $l_0 = 0.001$, $\omega = 2$, $\rho = 0.1$, and $\sigma = 0.0001$, (c) changes in σ for $R_1 = 0.1$, h = 0.0001, $\mu = 0.0001$, $\omega = 2$, and $\rho = 0.1$, and (d) changes in l_0 for $R_1 = 0.1$, h = 0.0001, $\omega = 2$, $\rho = 0.1$, and $\sigma = 0.001$.



FIGURE 9: Parameters affecting the superwave solution (39): (a) changes in μ for $R_1 = 0.01$, $\alpha = 0.8$, h = 0.006, k = 1, $l_0 = 0.001$, $\rho = 0.00002$, and $\sigma = 0.01$ and (b) changes in h for $R_1 = 0.01$, $\alpha = 0.8$, h = 0.006, k = 1, $l_0 = 0.001$, $\rho = 0.00002$, and $\sigma = 0.01$.



FIGURE 10: Graphic representation for the singular wave solution (15). (a) 3D graphic and (b) 2D contour.



FIGURE 11: Graphic representation for the singular wave solution (37). (a) 3D graphic and (b) 2D contour.

5. Conclusion

This work is aimed to study analytically the double-chain model for deoxyribonucleic acid (DNA). A certain wave transformation has been applied to equation (1) to transform it into an ordinary differential equation. The integration of this equation reacquired some studies on the parameters. This study has been performed by applying the complete discrimination of the polynomial $F_4(p)$. Moreover, we have determined the possible interval of real propagations. Such study is more significant because the missing of such study implies to loss some solutions and also, give rise to complex solutions which are undesirable in real problems. For instance, there are several solutions corresponding to the same conditions on the discriminant system as outlined in Case 4. We have introduced new waves' solutions for equation (1). Let us compare the results obtained in the present article

with the well-known results obtained by other authors using different methods as follows: our results in a new doublechain model of DNA are new and different from those obtained in references [19-24]. We have studied the influence of some parameters such as the distance between the two strands, the stiffness of the elastic membrane, the area of the cross section of each strand, and the height of the membrane in the equilibrium. We have considered two types of solutions: one is kink (24) and the other is periodic (28). We have shown graphically the amplitude of the kink solution is decreased when the distance between the two strands or the area of the cross section of each strand is increased, while it is increased when the stiffness of the elastic membrane or the height of the membrane in the equilibrium is increased. For more clarification, see Figure 6. The amplitude of the periodic solution remains approximately unchanged when these physical parameters are changed, but the width has been affected. The width is increased due to the increase of the distance between the two strands or the area of the cross section of each strand, while it is decreased as a result of increasing the stiffness of the elastic membrane or the height of the membrane in the equilibrium. For more illustrations, see Figure 10. From another point of view, this ODE has been expressed as a one-dimensional Hamiltonian system that describes the physical motion of a particle with one degree of freedom under the action of potential function V(p) given by (45). Based on the Hamiltonian concepts, we have studied some qualitative analyses such as phase portrait and bifurcation. The description of phase space has been presented through Propositions 1, 2, and 3. Moreover, these propositions contain the conditions for the existence of periodic and solitary wave solutions.

Data Availability

No data were used to support the study.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

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Research Article

Some Novel Analytical Approximations to the (Un)damped Duffing–Mathieu Oscillators

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Some novel exact solutions and approximations to the damped Duffing–Mathieu-type oscillator with cubic nonlinearity are obtained. This work is divided into two parts: in the first part, some exact solutions to both damped and undamped Mathieu oscillators are obtained. These solutions are expressed in terms of the Mathieu functions of the first kind. In the second part, the equation of motion to the damped Duffing–Mathieu equation (dDME) is solved using some effective and highly accurate approaches. In the first approach, the nonintegrable dDME with cubic nonlinearity is reduced to the integrable dDME with linear term having undermined optimal parameter (maybe called reduced method). Using a suitable technique, we can determine the value of the optimal parameter and then an analytical approximation is obtained in terms of the Mathieu functions. In the second approach, the ansatz method is employed for deriving an analytical approximation in terms of trigonometric functions. In the third approach, the homotopy perturbation technique with the extended Krylov–Bogoliubov–Mitropolskii (HKBM) method is applied to find an analytical approximation to the dDME. Furthermore, the dDME is solved numerically using the Runge–Kutta (RK) numerical method. The comparison between the analytical and numerical approximations is carried out. All obtained approximations can help a large number of researchers interested in studying the nonlinear oscillations in a plasma can be reduced to the family of Mathieu-type equation, Duffing-type equation, etc.

1. Introduction

Several physical and natural phenomena related to biology, chemistry, physics, engineering problems, and so on can be modelled by both ordinary differential equations (ODEs) and partial differential equations (PDEs) for studying the nonlinear self-excited oscillators [1–12]. Also, in many reallife problems, some internal and external forces that can affect the system under consideration cannot be neglected. For example, the friction and collisional force and many others that affect on the motion of particles, whether in solid, liquid, gas or plasma physics, cannot be neglected. Therefore, these forces must be included in the mathematical models that will be used for studying the natural and physical problem, such as investigating the nonlinear oscillations in various plasma models [13–23]. Interest in the study of nonlinear oscillations in a plasma is due to its many potential applications. Nowadays, plasma processing is seen as an important and effective technology which has been able to enter into all modern industries. In addition, plasma had a

great credit for the modern technology in electronics, medicine, agriculture, biomedicine, automobiles, optics, aerospace, telecommunications, solar energy, polymers, papers, textiles, etc. [16, 24]. Therefore, we focus our attention on the applications of the family of Duffing-type equation for modelling the nonlinear oscillations in a plasma. The Duffing-type oscillator is one of the most popular differential equations that has spread widely due to its ability to explain many nonlinear phenomena in various fields of science and in mechanical systems and engineering. This equation is a mathematical model described by a second order of ODEs with a nonlinear spring force. It is used for describing the motion of (un)damped oscillator rather than simple harmonic motion. Motivated by potential engineering and plasma physics applications in addition to many others applications in electronic circuits and microcontroller, the family of Duffing equation such as (un) damped Duffing equation with lower and higher-order nonlinearities, (un)damped Duffing-Helmholtz equation, and (un)damped Mathieu-Duffing equation have received wide attention due to their ability for investigating the mechanism of a rigid pendulum oscillator, oscillations in different plasma models, and so on. This family of secondorder differential equations has provided useful and successful models for investigating the nonlinear oscillations and chaotic nature. The biggest challenge in the study of the dynamics of nonlinear mechanical systems is to find some real solutions (including the analytical and numerical solutions) to the evolution equations that are used for describing the characteristics of nonlinear phenomena under consideration. Accordingly, studying the solutions of many equations of motion to various oscillators is one of the most difficult tasks facing many researchers.

The solutions associated to mentioned evolution equations and many other related equations have been studied extensively due to the fact that such equations arise in a variety of realistic problems. For instance, the fluid equations of electron-ion unmagnetized cold plasma were reduced to Mathieu equation in order to investigate the electron waves [13]. During this study, the authors assumed that the density perturbations of the plasma species are only time-dependent functions and do not depend on space. Also, the basic set of fluid equations to a multicomponent complex plasma consisting of inertial two types of dust grains (both positive and negative charges) as well as inertialess Maxwellian species including electrons and ions were reduced to a Mathieu-type equation for studying the excitation of dustacoustic oscillations [14]. Moreover, a nonlinear Van der Pol-Mathieu-type equation was derived for the dust grain density in order to investigate the dynamics of dust-acoustic oscillations in a dusty plasma consisting of inertial Boltzmann distributed species (electrons and ions) and inertialess dust grains [15]. A modified Van der Pol-Duffing oscillator with forced term was used in the study and was used in modelling the dynamics of nonlinear oscillations in different plasma models [16]. More recently, the multistage method was used for solving the damping Duffing equation with forced term in order to model the oscillations in a complex unmagnetized plasma [17].

Due to the importance of the family of Duffing-type equation and motivated by the mentioned studies, we focus our attention on the analysis of the so-called (un)damped Mathieu–Duffing oscillator with a twin-well potential [25].

$$\mathbb{R} \equiv \ddot{x} + (\alpha - Q_0 \cos(\Omega t))x + \beta x^3 = 0, \tag{1}$$

and

$$\mathbb{R}_1 \equiv \ddot{x} + \varepsilon \dot{x} + (\alpha - Q_0 \cos(\Omega t))x + \beta x^3 = 0, \qquad (2)$$

for studying the vibrating/oscillating behavior of systems described by (1) and (2), where $(\alpha, \beta) > 0$ are, respectively, the stiffness coefficients of the linear and nonlinear terms and ε represents the coefficients of damping term. The longitudinal loading is periodic, Ω and Q_0 are the frequency and excitation strength of the periodic loading, respectively. The total energy of the undamped Mathieu–Duffing oscillator (DMO) according to (1) is defined by $H = H_0 + H_1$, where $H_0 = 1/2\dot{x}^2 - 1/2\alpha x^2 +$ $1/4\beta x^4$ and $H_1 = -1/2x^2Q_0\cos(\Omega t)$ are, respectively, the unperturbed and perturbation in the Hamiltonian of (1). There is another form to the equation of periodic motion which is called (un)damped Mathieu–Helmholtz oscillator.

$$\begin{cases} \ddot{x} + (\alpha - Q_0 \cos(\Omega t))x + \beta x^2 = 0, \\ \ddot{x} + \varepsilon \dot{x} + (\alpha - Q_0 \cos(\Omega t))x + \beta x^2 = 0. \end{cases}$$
(3)

The analytical solution to the damped Mathieu–Helmholtz oscillator (3) was obtained using the finite Fourier series expansion [26]. Note that in evolution equation (3), the nonlinear term βx^2 is different from the nonlinear term in equations (1) and (2).

The objectives of our study are to find some novel solutions to the (un)damped Duffing-Mathieu-type oscillator, under the initial conditions x(0) = 0 and $x'(0) = \dot{x}_0$. Two cases for Duffing–Mathieu-type oscillator will be discussed. In the first case, we will get some exact solutions to (un)damped Mathieu equation in terms of the Mathieu functions of the first kind. In the second case, the damped Duffing-Mathieu oscillator (dDMO) will be solved analytically and numerically using some different approaches. In the first approach, the cubic nonlinear term in equation (1) βx^3 is replaced by the linear term $\beta \kappa x$, where the constant $\kappa \ge 0$ represents an optimal parameter. Then, the nonintegrable dDMO reduces to an integrable one which has an exact solution but with undermined parameter κ . Using a suitable technique, we can determine the value of the optimal parameter κ . Thus, we can get an analytical approximation to the dDMO (2) in terms of the Mathieu functions. For the second approach, the ansatz method with the help of the solution to the undamped Duffing oscillator is employed to derive an analytical approximation to the dDMO (2) in the form of trigonometric functions. Furthermore, the homotopy perturbation technique with the extended Krylov-Bogoliubov-Mitropolskii (KBM) which is called HKBM method is also devoted for solving the dDME (2) for arbitrary physical parameters [27, 28].

2. Mathematical Analysis

Here, we proceed to find some approximate solutions to both undamped and damped Duffing-Mathieu oscillators (1) and (2), respectively. Below we discuss the different approaches for solving the mentioned equations.

2.1. An Exact Solution to (Un)damped Mathieu Equation. Both undamped Mathieu equation, i.e., (1) for $\beta = 0$, and the damped Mathieu equation, i.e., (2) for $\beta = 0$, have exact solutions in the form of Mathieu functions. First, let us find the solution of the following damped Mathieu equation:

$$\ddot{x} + \varepsilon \dot{x} + (\alpha - Q_0 \cos(\Omega t))x = 0, \tag{4}$$

with subjected to the initial conditions (ICs): $x(0) = x_0$ and $x'(0) = \dot{x}_0$.

Using the following MATHEMATICA command

$$g[t_{-}] \coloneqq x''[t] + 2\varepsilon x'[t] - (\alpha - Q_0 \text{Cos}[\Omega t])x[t]$$

$$(\text{NDSolve}[g[t] == 0\&\&x[0] == x_0\&\&x'[0] == \dot{x}_0, x[t], t][1, 1, 2]//\text{FullSimplify})^1,$$
(5)

we can get the exact solution to (4) as follows:

$$x(t) = e^{-\varepsilon t} \left(\frac{x_0 M C_1(t,\varepsilon)}{M C_1(0,\varepsilon)} + \frac{2(\varepsilon x_0 + \dot{x}_0)}{\Omega} \frac{M S_1(t,\varepsilon)}{\text{MathieuSPrime}[4(\alpha - \varepsilon^2)/\Omega^2, 2Q_0/\Omega^2, 0]} \right),$$
(6)

with

$$MC_{1}(t,\varepsilon) = \text{MathieuC}\left[\frac{4(\alpha-\varepsilon^{2})}{\Omega^{2}}, \frac{2Q_{0}}{\Omega^{2}}, \frac{\Omega}{2}t\right],$$

$$MS_{1}(t,\varepsilon) = \text{MathieuS}\left[\frac{4(\alpha-\varepsilon^{2})}{\Omega^{2}}, \frac{2Q_{0}}{\Omega^{2}}, \frac{\Omega}{2}t\right],$$
(7)

where MathieuS and MathieuC are the Mathieu functions of the first kind or sometimes called sine-elliptic and cosineelliptic, respectively. For $\varepsilon = 0$, the damped Mathieu equation (4) reduces to the following undamped Mathieu equation:

$$\ddot{x} - (\alpha + Q_0 \cos(\Omega t))x = 0, \qquad (8)$$

and solution (6) reduces to the following one:

$$x(t) = \frac{x_0 M C_1(t,0)}{M C_1(0,0)} + \frac{2\dot{x}_0 M S_1(t,0)}{\Omega \text{MathieuSPrime} \left[4\alpha/\Omega^2, 2Q_0/\Omega^2, 0\right]}.$$
(9)

The exact solution (6) is compared with RK numerical solution as shown in Figures 1(a) and 1(b) for different values of x_0 .

2.2. Some Analytical Approximations to the Damped Duffing-Mathieu Oscillator. Here, we proceed to discuss two techniques (the hybrid p-expansion method and the ansatz method) for finding some analytical approximations to dDMO (2). For studying dDMO (2), first we rewrite equation (2) in the following new i.v.p.

$$\begin{cases} \mathbb{R}_1 = 0, \\ x(0) = x_0 \& x'(0) = \dot{x}_0. \end{cases}$$
(10)

2.2.1. First Approach: Reduced Method. For $\beta \neq 0$ and $\alpha > 0$, we may obtain simple approximation to the i.v.p. (10) by replacing the cubic term βx^3 by the linear term $\beta \kappa x$, where the constant $\kappa \ge 0$ which is used as an optimal parameter to reduce the residual error. Accordingly, dDMO (2) of cubic nonlinearity reduces to the following dDMO with linear term. Thus, we can replace the i.v.p. (10) by the following new i.v.p.

$$\begin{cases} \ddot{x} + 2\varepsilon \dot{x} + (\alpha - Q_0 \cos(\Omega t))x + \beta \kappa x = 0, \\ x(0) = x_0 \text{ and } x'(0) = \dot{x}_0. \end{cases}$$
(11)

Thus, the exact solution to the i.v.p. (11) is expressed by

$$x_{\kappa} \equiv x_{\kappa}(t) = e^{-\varepsilon t} \left(\frac{x_0 M C_2(t,\varepsilon,\kappa)}{M C_2(0,\varepsilon,\kappa)} + \frac{2(\varepsilon x_0 + \dot{x}_0)}{\Omega} \frac{M S_2(t,\varepsilon,\kappa)}{\text{MathieuSPrime}\left[4(\beta \kappa - \varepsilon^2 + \alpha)/\Omega^2, 2Q_0/\Omega^2, 0\right]} \right),$$
(12)



FIGURE 1: Both exact solution (6) and RK numerical solution to the damped Mathieu equation (4) are compared with each other for different values to the initial amplitude x_0 .

with

$$MC_{2}(t,\varepsilon,\kappa) = \text{MathieuC}\left[\frac{4(\beta\kappa-\varepsilon^{2}+\alpha)}{\Omega^{2}}, \frac{2Q_{0}}{\Omega^{2}}, \frac{\Omega}{2}t\right],$$

$$MS_{2}(t,\varepsilon,\kappa) = \text{MathieuS}\left[\frac{4(\beta\kappa-\varepsilon^{2}+\alpha)}{\Omega^{2}}, \frac{2Q_{0}}{\Omega^{2}}, \frac{\Omega}{2}t\right].$$
(13)

The residual is defined as

$$R_{\kappa}(t) = \ddot{x}_{\kappa} + \varepsilon \dot{x}_{\kappa} + (\alpha - Q_0 \cos(\Omega t)) x_{\kappa} + \beta x_{\kappa}^3.$$
(14)

A suitable value of κ can be obtained by solving the equation $R_{\kappa}(t_0) = 0$ for some $t_0 > 0$, say $t_0 = 1$. Making use of the Padé approximate technique, for $0 < t_0 \le 1$ and $x_0 \approx 0$, we get

$$\kappa_{\text{suitable}} = \frac{Y_1}{Y_2},\tag{15}$$

with

$$Y_{1} = 3x_{0}^{2} \Big(Q_{0}t_{0}x_{0}^{2} - \alpha t_{0}x_{0}^{2} - 2\varepsilon t_{0}\dot{x}_{0}x_{0} - 4t_{0}\dot{x}_{0}^{2} - 2\dot{x}_{0}x_{0} \Big) \\ \times \Big(4Q_{0}t_{0}x_{0}^{2} - 4\alpha t_{0}x_{0}^{2} + 3\beta t_{0}x_{0}^{4} - 8\varepsilon t_{0}\dot{x}_{0}x_{0} - 6t_{0}\dot{x}_{0}^{2} - 8\dot{x}_{0}x_{0} \Big), \\ Y_{2} = -26\alpha Q_{0}t_{0}^{2}x_{0}^{4} + 12\beta Q_{0}t_{0}^{2}x_{0}^{6} - 52\varepsilon Q_{0}t_{0}^{2}\dot{x}_{0}x_{0}^{3} + 13Q_{0}^{2}t_{0}^{2}x_{0}^{4} \\ - 52Q_{0}t_{0}\dot{x}_{0}x_{0}^{3} - 30Q_{0}t_{0}^{2}\dot{x}_{0}^{2}x_{0}^{2} + 13\alpha^{2}t_{0}^{2}x_{0}^{4} - 12\alpha\beta t_{0}^{2}x_{0}^{6} \\ + 52\alpha\varepsilon t_{0}\dot{x}_{0}x_{0}^{3} + 52\alpha t_{0}\dot{x}_{0}x_{0}^{3} + 30\alpha t_{0}^{2}\dot{x}_{0}^{2}x_{0}^{2} + 9\beta^{2}t_{0}^{2}x_{0}^{8} \\ - 24\beta\varepsilon t_{0}^{2}\dot{x}_{0}x_{0}^{5} - 24\beta t_{0}\dot{x}_{0}x_{0}^{5} + 12\beta t_{0}^{2}\dot{x}_{0}^{2}x_{0}^{4} + 52\varepsilon^{2}t_{0}^{2}\dot{x}_{0}^{2}x_{0}^{2} \\ + 104\varepsilon t_{0}\dot{x}_{0}^{2}x_{0}^{2} + 60\varepsilon t_{0}^{2}\dot{x}_{0}^{3}x_{0} + 60t_{0}\dot{x}_{0}^{3}x_{0} + 12t_{0}^{2}\dot{x}_{0}^{4} + 52\dot{x}_{0}^{2}x_{0}^{2}.$$
(16)

For a given κ , the residual error $L_R(\kappa)$ of the approximation (12) to the i.v.p. (10) is defined as

$$L_{R}(\kappa) = \max_{0 \le t \le T} |R_{\kappa}(t)| = \max_{0 \le t \le T} |\ddot{x}_{\kappa} + 2\varepsilon \dot{x}_{\kappa} + (\alpha - Q_{0}\cos(\Omega t))x_{\kappa} + \beta x_{\kappa}^{3}|.$$
(17)

The optimal value of κ_{optimal} for the parameter κ on the interval $0 \le t \le T$ is defined as

$$\kappa_{\text{optimal}} = \min_{\kappa > 0} L_R(\kappa). \tag{18}$$

Let us apply the obtained approximation (12) for investigating the properties of the damping oscillations to the dDMO (2) at different values to the physical parameters $(\alpha, \beta, \varepsilon, \Omega, Q_0, x_0, \dot{x}_0)$. The profile of the approximation (12) using the values of κ_{suitable} (given in equation (15)) and κ_{optimal} (given in (18)) is displayed in Figures 2 and 3 at $(\alpha, \beta, \varepsilon, \Omega, Q_0, x_0, \dot{x}_0) = (4, 1, 0.1, 0.5, 0.1, 0, 0.2)$ and $(\alpha, \beta, \varepsilon, \Omega, Q_0, x_0, \dot{x}_0) = (4, 1, 0.1, 0.5, 0.1, \pi/6, 0.2)$, respectively. The obtained results showed that this approximation gives results with good and acceptable accuracy.

2.2.2. The Ansatz Method for Solving dDMO. Now, we can summarize the main points to get some approximations to the i.v.p. (10) in the following steps.

Step 1. Let us assume the following ansatz:

$$x(t) = y(f(t)),$$
 (19)

where the time-dependent function $f \equiv f(t)$ can be determined later and the function $y \equiv y(t)$ represents the solution of the following i.v.p.

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FIGURE 2: The profile of the approximation (12) using the values of κ_{suitable} (given in equation (15)) and κ_{optimal} (given in (18)) is plotted against t for $(\alpha, \beta, \varepsilon, \Omega, Q_0, x_0, \dot{x}_0) = (4, 1, 0.1, 0.5, 0.1, 0, 0.2)$.



FIGURE 3: The profile of the approximation (12) using the values of κ_{suitable} (given in equation (15)) and κ_{optimal} (given in (18)) is plotted against *t* for $(\alpha, \beta, \varepsilon, \Omega, Q_0, x_0, \dot{x}_0) = (4, 1, 0.1, 0.5, 0.1, \pi/6, 0.2).$

$$\begin{cases} \ddot{y} + 2\varepsilon \dot{y} + (\alpha - Q_0)y + \beta y^3 = 0, \ y(0) = x_0 \& y'(0) = \dot{x}_0. \end{cases}$$
(20)

Step 2. Inserting ansatz (19) into i.v.p. (10), we get

$$\mathbb{R}_{1} = y'(f) \Big(-2\varepsilon f'^{2} + 2\varepsilon f' + f'' \Big) + y(f) \Big(\alpha - Q_{0} \cos(\Omega t) + f'^{2} (Q_{0} - \alpha) \Big)$$
(21)
+ $\beta y^{3}(f) \Big(1 - f'^{2} \Big).$

Step 3. For vanishing the coefficient of y(f) in equation (21), we get

$$f'^{2} = \frac{-\alpha + Q_{0} \cos{(\Omega t)}}{(Q_{0} - \alpha)}.$$
 (22)

Step 4. Integrating equation (22) and with the help of f(0) = 0, we have

$$f(t) = \frac{2}{\Omega} E\left(\frac{\Omega}{2}t, \frac{2Q_0}{Q_0 - \alpha}\right),\tag{23}$$

where *E* stands for the EllipticE function.

Now, the value of f(t) has been determined but the solution of the i.v.p. (20) needs to be determined. Thus, we are faced with two things: either we use one of the solutions found in the literature [29] or try to find another solution in the form of trigonometric functions.

Step 5. In this step, we proceed to find a solution to i.v.p. (20) in the form of trigonometric functions. Without loss of generality, i.v.p. (20) can be redefined as

$$\begin{cases} \mathbb{R}_2 = \ddot{\nu} + 2\varepsilon\dot{\nu} + p\nu + q\nu^3 = 0, \\ \nu(0) = \nu_0 & \varepsilon'(0) = \dot{\nu}_0, \end{cases}$$
(24)

where i.v.p. (24) is the same as i.v.p. (20) with $p = (\alpha - Q_0)$, $q = \beta$, and v(t) = y(t).

Step 6. Our objective here is to derive another solution that does not involve elliptic functions but an elementary solution. To do that, we assume $\varepsilon > 0$, and for $\lim_{t \to \infty} v(t) = 0$, we get

$$v(t) = c_0 e^{-\rho t} \cos\left(h(t) + \arccos\left(\frac{x_0}{c_0}\right)\right), \tag{25}$$

where h(0) = 0 and h(t) is undermined function.

Step 7. By substituting solution (25) into i.v.p. (24), we have $\mathbb{R}_{2} = c_{0} \sin(\theta) e^{-\rho t} \left(-2\varepsilon h' + 2\rho h' - h''\right) + \frac{1}{4} c_{0}^{3} q \cos(3\theta) e^{-3\rho t} + \frac{1}{4} c_{0} \cos(\theta) e^{-3\rho t} \left[3c_{0}^{2} q + 4e^{2\rho t} \left(-2\varepsilon \rho - h'^{2} + p + \rho^{2}\right)\right].$ (26)

Step 8. For vanishing the coefficient of $\cos(\theta)$ in equation (26): $3c_0^2q + 4e^{2\rho t}(-2\varepsilon\rho - {h'}^2 + p + \rho^2) = 0$, we have

$$h' = \pm \frac{1}{2}\sqrt{4p - 8\varepsilon\rho + 4\rho^2 + 3c_0^2 q e^{-2\rho t}},$$
 (27)

and by solving (27) with h(0) = 0, we get

$$h(t) = H(t) - H(0),$$
 (28)

with

$$H(t) = \frac{1}{\rho} \left(\sqrt{\Pi} \tanh^{-1} \left(\sqrt{1 + \frac{3c_0^2 q e^{-2\rho t}}{4\Pi}} \right) - \sqrt{\frac{3}{4}c_0^2 q e^{-2\rho t} + \Pi} \right),$$

$$H(0) = \frac{1}{\rho} \left(\sqrt{\Pi} \tanh^{-1} \left(\sqrt{1 + \frac{3c_0^2 q}{4\Pi}} \right) - \sqrt{\frac{3}{4}c_0^2 q + \Pi} \right),$$

(29)

where $\Pi = (p - 2\varepsilon \rho + \rho^2)$.

Step 9. The number c_0 is obtained from the condition $v'(0) = \dot{v}_0$ and it is a solution to the quartic

$$3qc_0^4 + (4p - 8\varepsilon\rho + 4\rho^2 - 3qv_0^2)c_0^2 - 4(pv_0^2 - 2\varepsilon\rho v_0^2 + 2\rho^2 v_0^2 + 2\rho v_0 \dot{v}_0 + \dot{v}_0^2) = 0,$$
(30)

where the number ρ is a free/optimal parameter that is chosen in order to minimize the residual error. Its default value is $\rho = \varepsilon$.

Step 10. Finally, the trigonometric approximation to i.v.p. (10) is obtained:

$$x(t) = y(f(t)) = c_0 e^{-\rho f(t)} \cos\left(h(f(t)) + \arccos\left(\frac{x_0}{c_0}\right)\right).$$
(31)

Step 11. Also, we can solve i.v.p. (20) using RK numerical method and then replacing $t \longrightarrow f(t)$ (given in equation (23)). The following MATHEMATICA command is introduced for this purpose:

$$g[t_{-}] \coloneqq y''[t] + 2\varepsilon y'[t] + (\alpha - Q_{0}) y[t] + \beta y[t]^{3};$$

$$RK[t_{-}] \coloneqq \text{NDSolve}[g[t] == 0 \& \& y[0] == x_{0} \& \& y'[0]$$

$$= \dot{x}_{0}, y[t], t][1, 1, 2],$$
(32)

$$x[t_{-}] \coloneqq RK[f[t]]. \tag{33}$$

Both analytical and numerical approximations (31) and (33) to i.v.p. (10) are, respectively, plotted against the RK numerical solution as illustrated in Figures 4 and 5. Also, at $(\alpha, \beta, \varepsilon, \Omega, Q_0, \dot{x}_0) = (4, 1, 0.1, 0.5, 0.1, 0.2)$, the maximum global distance of both approximations (31) and (33) is estimated for different values to x_0 as

$$L_{d}(x_{0} = 0) = \max_{0 \le t \le 30} \left| RK - x(t)_{\text{Approx.}(26)} \right| = 0.000506901,$$

$$L_{d}(x_{0} = 0) = \max_{0 \le t \le 30} \left| RK - x(t)_{\text{Approx.}(28)} \right| = 0.000482946,$$

$$L_{d}\left(x_{0} = \frac{\pi}{6}\right) = \max_{0 \le t \le 30} \left| RK - x(t)_{\text{Approx.}(26)} \right| = 0.00415468,$$

$$L_{d}\left(x_{0} = \frac{\pi}{6}\right) = \max_{0 \le t \le 30} \left| RK - x(t)_{\text{Approx.}(28)} \right| = 0.00266014.$$
(34)

The obtained results show the high accuracy and efficiency of the obtained approximations (31) and (33). Moreover, these approximations are stable against long time and for arbitrary values of the physical parameters.

2.3. The Homotopy Extended Krylov-Bogoliubov-Mitropolskii Method. The homotopy extended Krylov-Bogoliubov-Mitropolskii (HKBM) method may be used for solving both conservative and nonconservative oscillators. Based on this method (more details can be found in [27, 28]), i.v.p. (10) can be redefined as

$$\begin{cases} \ddot{x} + (\alpha - Q_0)x + p[\varepsilon \dot{x} + Q_0 (1 - \cos(\Omega t))x + \beta x^3] = 0, \\ x(0) = x_0 \text{ and } x'(0) = \dot{x}_0, 0 \le t \le T, \end{cases}$$
(35)

where $x_p \equiv x_p(t)$ indicates the solution of i.v.p. (35) while the solution of the dDMO (2) is obtained for p = 1. For $\omega_0 = \sqrt{\alpha - Q_0}$, $\phi(t) = Q_0(1 - \cos(\Omega t))$ and $\alpha - Q_0 > 0$, i.v.p. (35) can be written in the following reduced form:



FIGURE 4: Both trigonometric approximation (31) and RK numerical simulation to the damped Duffing–Mathieu problem (10) are compared with each other for different values to the initial amplitude x_0 .



FIGURE 5: Both RK numerical simulation (32) and RK numerical simulation (33) using the definition of $t \rightarrow f(t)$ to the damped Duffing-Mathieu problem (10) are compared with each other for different values to the initial amplitude x_0 .

$$\begin{cases} \ddot{x} + \omega_0^2 x + p(\varepsilon \dot{x} + \phi(t)x + \beta x^3) = 0, \\ x(0) = x_0 \text{ and } x'(0) = \dot{x}_0, 0 \le t \le T. \end{cases}$$
(36)

According to the HKBM method, the following ansatz solution is introduced:

$$x_{p} = a \cos(\psi) + \sum_{n=1}^{N} p^{n} u_{n}(a, \psi) + O(p^{N+1}), \qquad (37)$$

where each $u_n \equiv u_n(a, \psi)$ is a periodic function in ψ , and both amplitude *a* and phase ψ are assumed to vary with time and subject to the conditions

$$\frac{da}{dt} \equiv \dot{a} = \sum_{n=1}^{N} p^{n} A_{n}(a) + O(p^{N+1}),$$

$$\frac{d\psi}{dt} \equiv \dot{\psi} = \omega_{0} + \sum_{n=1}^{N} p^{n} \psi_{n}(a) + O(p^{N+1}),$$
(38)

where $a \equiv a(t)$ and $\psi \equiv \psi(t)$.

Inserting ansatz solution (37) and using (38) and after several tedious calculations, we can determine the unknown time-dependent functions (u_n, ψ_n, A_n, a) . To avoid the socalled secularity, we choose only the solution that does not contain $\cos \psi$ nor $\sin \psi$. For N = 1, we get



FIGURE 6: Both HKBM first-order approximate solution (42) and RK numerical simulation are plotted against different values of the initial angle x_0 .



FIGURE 7: Both HKBM first-order approximate solution (42) and RK numerical simulation are plotted against different values of the damping parameter ε .

$$\dot{a} = -\varepsilon a(t), \qquad a = c_0 e^{-\varepsilon t},$$

$$\dot{\psi} = \frac{3\beta a(t)^2 + 4\phi(t)}{8\omega_0} + \omega_0, \qquad (39) \qquad \psi = \frac{e^{-2\varepsilon t}}{16\varepsilon \Omega \omega_0} \left[8\varepsilon e^{2\varepsilon t} \left(2c_1 \Omega \omega_0 - Q_0 \left(\Omega t + \sin\left(\Omega t\right)\right) + 2\alpha \Omega t\right) + 3\beta c_0^2 \Omega \left(e^{2\varepsilon t} - 1\right)\right].$$

$$u_1(a, \psi) = \frac{a^3 \beta \cos(3\psi)}{32\omega_0^2}, \qquad (41)$$

p = 1:

and

$$x_{p}(t) = a \cos(\psi) + p \frac{\beta}{32\omega_{0}^{2}} a^{3} \cos(3\psi).$$
 (40)

By solving system (39), we have

$$x(t) = x_1(t) = a \cos(\psi) + \frac{\beta}{32\omega_0^2} a^3 \cos(3\psi),$$
(42)

The first-order approximate solution is obtained for

where the values of (a, ψ) are defined in (41) while the constants c_0 and c_1 can be obtained from the initial conditions.

The comparison between the HKBM first-order approximate solution (37) and the RK numerical simulation is reported as shown in Figures 6(a) and 6(b) for $x_0 = 0$ and $x_0 = \pi/6$, respectively. Also, both HKBM first-order approximate solution (42) and RK numerical simulation are, respectively, compared with each other for weak ($\varepsilon = 0.1$) and strong ($\varepsilon = 0.5$) damping as illustrated in Figures 7(a) and 7(b). Furthermore, at $(\alpha, \beta, \Omega, Q_0, \dot{x}_0) = (4, 1, 0.5, 0.1, 0.2)$ and for different values to (x_0, ε) , the maximum global distance error to the HKBM first-order approximate solution (42) is estimated as

$$\begin{split} L_d \left(x_0 = 0 \right) &= \max_{0 \le t \le 30} \left| RK - x \left(t \right)_{\text{HKBM}(37)} \right| = 0.00110165, \\ L_d \left(x_0 = \frac{\pi}{6} \right) &= \max_{0 \le t \le 30} \left| RK - x \left(t \right)_{\text{HKBM}(37)} \right| = 0.0061929, \\ L_d \left(\varepsilon = 0.1 \right) &= \max_{0 \le t \le 30} \left| RK - x \left(t \right)_{\text{HKBM}(37)} \right| = 0.00110165, \\ L_d \left(\varepsilon = 0.5 \right) &= \max_{0 \le t \le 30} \left| RK - x \left(t \right)_{\text{HKBM}(37)} \right| = 0.00493896. \end{split}$$

$$(43)$$

It is clear that the HKBM first-order approximate solution (42) is characterized by high accuracy and more stability at long time.

3. Conclusion

Given the importance of nonlinear oscillations in plasma physics and engineering and their strong connection to the family of the Duffing-type oscillator, in this work, some exact solutions to the damped and undamped Mathieu equations as well as some analytical approximations to the damped Duffing-Mathieu oscillator (dDMO) using different approaches have been obtained. The exact solutions to both damped and undamped Mathieu equation have been obtained in the terms of Mathieu functions of the first kind. These solutions are numerically compared with the Runge-Kutta (RK) numerical simulation. It was observed that both exact and numerical solutions are completely matched with each other in the whole time interval. On the other hand, the dDMO has been solved using some different approaches. In the first one, the nonintegrable dDMO with cubic nonlinear term (βx^3) has been reduced to an integrable dDMO with linear term $(\beta \kappa x)$ in which κ is undermined optimal parameter. The kappa optimal parameter κ has been determined using a suitable technique as we discussed in the text above. After determining the kappa optimal parameter, a highly accurate analytical approximation has been obtained in terms of the Mathieu functions. In the second approach, a highly accurate analytical approximation has been derived in detail in terms of trigonometric functions using the ansatz method. In the third technique, the homotopy extended Krylov-Bogoliubov-Mitropolskii (HKBM) method was used for

getting an effective analytical approximation. Furthermore, the dDMO has been analyzed numerically using the RK numerical method. The comparison between all obtained approximations and the RK numerical solutions has been carried out. Moreover, the maximum global distance error in the whole time interval to all obtained approximations has been estimated. All obtained approximations are characterized by the high accuracy and efficiency in addition to being more stable for a long time.

3.1. Future Work. We may solve the following oscillators using of the methods described in this paper:

3.1.1. Future Idea I. Cubic-quintic Duffing–Mathieu equation:

$$\begin{cases} \ddot{x} + \omega_0^2 x + 2\varepsilon \dot{x} + \phi(t)x + \beta x^3 + \gamma x^5 = 0, \\ x(0) = x_0 \& x'(0) = x_0, \quad 0 \le t \le T. \end{cases}$$
(44)

3.1.2. Future Idea II. Forced damped Duffing–Mathieu equation:

$$\begin{cases} \ddot{x} + \omega_0^2 x + 2\varepsilon \dot{x} + \phi(t)x + \beta x^3 = F(t), \\ x(0) = x_0 \& x'(0) = x_0, \quad 0 \le t \le T. \end{cases}$$
(45)

3.1.3. Future Idea III. The forced Van der Pol-Duffing oscillator:

$$\begin{cases} \ddot{x} - \varepsilon (1 - x^2) \dot{x} + \omega_0^2 x + \beta x^3 = F(t), \\ x(0) = x_0 \& x'(0) = x_0, \quad 0 \le t \le T, \end{cases}$$
(46)

and many others oscillators.

Data Availability

The data generated or analyzed during this study are included within the article.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

Authors' Contributions

All authors contributed equally to this study and approved the final version of the manuscript.

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Review Article

Dynamics of Chinese Export Comparative Advantage: Analysis Based on RSCA Index

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Based on RSCA index, using kernel density estimation, Markov chain transition probability matrix, and survival analysis method, this paper analyzes the dynamics of Chinese export comparative advantage from 2001 to 2020 and draws the following conclusions. Firstly, after 20 years of export trade development, although the comparative advantage of a few commodities of China has weakened and comparative disadvantage has increased, the comparative advantage of most commodities is improving, and the overall distribution of comparative advantage remained unchanged. Secondly, the stability of Chinese comparative advantage is higher than liquidity, and liquidity as a whole shows a good trend. In addition, the viability of Chinese commodities with comparative advantage has performed well in the past 20 years. Therefore, China should optimize export mode based on comparative advantage.

1. Introduction

The export mode of country and region has long been one of the research hot-spots in the field of international trade. Although the new trade theory in the 1980s attributed the emergence of trade to two cornerstones, namely, comparative advantage and economies of scale, comparative advantage is still the main theoretical explanation [1]. According to David Ricardo's comparative advantage theory, the difference of relative labor productivity among countries leads to the difference of relative production cost, which leads to the difference of relative export price. Hence, a country should export commodities with comparative advantage and import commodities with comparative disadvantage. Regional relative labor productivity will not always be in a static state, but will continue to change with the passage of time, and then the regional comparative advantage will also change. Therefore, comparative advantage is a dynamic concept and develops endogenously over time [2]. Countries and regions continuously strengthen or weaken the original comparative advantage due to factor

endowment, technological progress, industrial policy, and so on and even lead to the reversal of original comparative advantage. In this way, the dynamics of comparative advantage is not only an indirect reflection of the changes of regional factor endowment and technology level, but also an important content to measure the impact of government policies. Hence, how to measure the dynamics of comparative advantage naturally becomes the initial task of researchers.

Since implementation of the strategy of reform and opening up in the late 1970s, China's economy has been fully integrated into the process of economic globalization. Chinese government has vigorously developed open economy, practiced open economic system and mechanism, deepened foreign trade and investment policies adapted to its national conditions, actively developed bilateral and multilateral trade relations, integrated multilateral trade organizations, implemented "going global" strategy, and deepened "The Belt and Road" initiative. These measures have improved Chinese foreign trade development environment, trained a large number of various ownership business entities facing the world market, effectively optimized export commodity structure, and improved international competitiveness. After accession to WTO, although China has experienced the 2008 international financial crisis, trade friction with major countries, rising domestic production cost, fluctuation of exchange rate, COVID-19 virus, etc., its export trade has made remarkable achievements. Chinese commodity exports amounted to US \$509.6 billion in 2001. By 2020, its exports reached \$2590.6 billion. The export scale has increased more than five times. China has become the largest commodity exporter and the largest foreign trade country in the world, and its import and export trade has become an important engine of global economic growth. With the growth of export scale, China's export commodity structure has also been significantly optimized, gradually reversing the export commodity structure dominated by primary products and labor-intensive products. In 2020, the export volume of manufactured commodities accounted for 95.5% of total exports volume in China, while primary products accounted for only 4.5%. In manufactured commodities, the percentage of mechanical and electrical products and high-tech products representing high technological level in total export is becoming higher and higher. For example, machinery and transportation equipment accounts for 48.6% in 2020, in such commodities, mechanical and electrical products and accessories, telephone communication and audio products, office machinery, and automatic data processing equipment are the main export commodity categories. The optimization of export commodity structure itself is the favorable result of dynamics of export comparative advantage in China. Therefore, it is necessary to conduct in-depth research on the dynamics of Chinese export comparative advantage.

The rest of this paper is organized as follows. The second part summarizes empirical methods of dynamics of comparative advantage, including a brief discussion of comparative advantage index and the statistic methods of index. The third part is the index and research methods used in this paper, including RSCA index and kernel density estimation, Markov chain transition probability matrix, and survival analysis methods. The fourth part presents results of empirical research. The fifth part is main conclusions and suggestions.

2. Literature Review

Analyzing change characteristics of comparative advantage index in certain period is the basic research way of dynamics of comparative advantage. This includes two interrelated aspects of the choice of index and application of statistical methods.

In terms of comparative advantage index, scholars have put forward various types of index since the 1950s, among which Balassa revealed that comparative advantage index (RCA) is most famous [3], but the index is also controversial. The main controversy is that RCA index has inherent defects in both theoretical basis and empirical application. For example, the mean value of RCA index is unstable and its distribution is nonnormal, so the accuracy of measuring comparative advantage is questionable. For this reason, later scholars put forward many alternative indexes with the aim of overcoming one or more shortcomings of the original RCA index. For example, Michaely put forward Michaely Index (MI) [4], Vollrath proposed relative trade advantage (RTS), relative export advantage (RC), and revealed competitiveness (In RCA) [5], Lafay proposed Lafay Index [6], Dalum and Laursen offered revealed symmetric comparative advantage index (RSCA) [7], Heon and Oosterhaven proposed additive (aggregated) revealed comparative advantage index (ARCA) [8], Cai and Yu proposed net export-revealed comparative advantage index (NRCA) [9], Wosiek and Visvizi proposed visviz wosiek RCA index (VWRCA) [10], and Andrey and Vladimir proposed new net trade index (nt RCA) [11]. These alternative indexes can sometimes alleviate some defects of RCA index in some specific cases. However, as Sanidas and Shin said, there is no perfect index [12].

Application of statistical methods on dynamics of comparative advantage is becoming more and more diverse and complex. For example, Benedicits and Tamberi used cumulative distribution, kernel density estimation, Lorentz curve, location index, and other methods based on RCA index [13]. Proudman and Redding and Hinloopen and Marrewijk used Markov chain transition probability matrix of RCA index and liquidity index methods [14, 15]. Laursen and Michele Alessandrini conducted regression analysis of RSCA index and Lafay Index, respectively [16, 17]. Bojnec and Fertö [2] and Olivera kostoska [18] also used regression, Markov chain transition probability matrix and survival analysis method of RCA index.

Application of the above indexes and statistical methods can reflect dynamics of regional comparative advantage to a certain extent; however, the following problems cannot be avoided. Firstly, the choice of index remains unsolved. Some scholars believe that the indexes based on the supply dimension, that is, only export data and no import data, lead to incomplete comparative advantage analysis [19]. Nevertheless, the author believes that if the demand dimension is considered, that is to say, the indexes including import data are adopted, the measurement is more distorted relative to supply dimension due to the influence of government policies, trade relations, and geographical factors. Although these factors can be measured separately, it is actually so challenging. In addition, it is reasonable to use the indexes based on the supply dimension in the current trade environment. After all, export is less affected by trade policies and trade relations than import. Of course, even if the problems related index selection is solved, there are still other straits. For example, using "ex post" trade data to reflect "ex ante" comparative advantage is naturally flawed [11, 20]. Therefore, these all depend on the progress of follow-up index research. Secondly, if we carefully study the specific methods related to dynamics of comparative advantage, we can find that most methods compare the index distribution of discrete time, such as between start time and end time, and ignore complete trend. In addition, the classification level of commodity also has an impact on the research results. For example, according to SITC classification, there may be a phenomenon that commodities with one-digit classification do not have comparative advantage, while commodities with two-digit or three-digit classification may have comparative advantage, and there may also be a phenomenon that commodities with one-digit classification have comparative advantage, but commodities with two-digit or three-digit classification may not have comparative advantage. These are also not conducive to the accurate analysis of comparative advantage. Hence, study on dynamics of comparative advantage may be more accurate when the commodity classification is more detailed. In view of this, referring to SITC three-digit classification, this paper selects RSCA index to carry out research on dynamics of export comparative advantage in China from 2001 to 2020. In addition, this study not only compares discrete years, but also studies the whole trend.

3. Methodology and Data

3.1. RSCARSCA Index, Commodity Classification, and Data

3.1.1. RSCA Index. Revealed symmetric comparative advantage (RSCA) index was proposed by Dalum and Laursen [7]. This index is the modification of the revealed comparative advantage (RCA) index. The RCA index formula is

$$RCA_{ij} = \frac{X_{ij} / \sum_{i=1}^{m} X_{ij}}{X_{in} / \sum_{i=1}^{m} X_{in}},$$
(1)

where X_{ij} is export volume of region j commodity i. $\sum_{i=1}^{m} X_{ij}$ is total export volume of region j commodity i. X_{in} is export volume of commodity i in reference region n. Then $\sum_{i=1}^{m} X_{in}$ is total export volume of reference region. The value range of RCA_{ij} index is $[0, +\infty)$. When the value of RCA_{ij} index is greater than 1, it indicates that region j has comparative advantage in commodity i; otherwise it is the opposite; when the value of index is equal to 1, it indicates the median point of comparative advantage.

RSCARSCA index is converted from RCA index to alleviate the defect of asymmetric distribution of RCA index. The formula is

$$RSCA_{ij} = \frac{RCA_{ij} + 1}{RCA_{ij} + 1},$$
(2)

where the value of RSCA_{*ij*} index ranges from [-1, 1]. When RSCA_{*ij*} value is greater than 0, it indicates that region *j* has comparative advantage in commodity *i*; otherwise it is the opposite. When the value of index is equal to 0, it represents the median point of comparative advantage and neutral comparative advantage.

3.1.2. Commodity Classification and Data. This paper quotes the three-digit commodity classification in the Standard of International Trade Classification (SITC Rev 3). The classified export data of China and world involved in the index calculation are all from UN COMTRADE database. Due to lack of world's three-digital classified commodity export data in the database, the annual export volume of three digit commodities of each country included in the database is aggregated as the total classified export volume of world. The export data of Chinese classified commodity from 2001 to 2020 are relatively complete in the database, and only two categories of commodities are not included in the analysis due to lack of data of complete years, that is, Ores and concentrates of uranium or thorium (286) and Gold, nonmonetary (excluding gold ores and concentrates (971). A total of 255 commodities are selected finally. In addition, the reason why RSCA index is selected is also due to the consideration of data quality. The quality of export data in the database is better than that of import data. If the selected comparative advantage index contains import data, it will be difficult to have measurement results of 255 categories due to research cycle, commodity classification level, and other reasons.

3.2. Dynamics of Comparative Advantage Method

3.2.1. Kernel Density Estimation. Kernel density estimation is a nonparametric method to estimate the probability density function of continuous random variables without assuming the basic distribution of random variables. Let $(x_1, x_2, ..., x_n)$ be a random sample from the same unknown probability density function f(x) and its kernel density estimator is

$$f(x) = \frac{1}{n} \sum_{i=1}^{N} K_h(x - x_i)$$

= $\frac{1}{nh} \sum_{i=1}^{N} K\left(\frac{x - x_i}{h}\right),$ (3)

where *K* is kernel function and h > 0 is the smoothing parameter (also called the bandwidth). In this paper, the default Epanechnikov kernel function of Stata software is used to obtain the kernel density curve.

After kernel density estimation, two-tailed Wilcoxon signed rank test is also performed. This test is a nonparametric test used to test the difference in the distribution of two samples. The premise of two samples is not independent, or matched samples or paired samples, or repeated measurement of a single sample. Thus, the Wilcoxon signed rank test tests for the null hypothesis of equal distributions through equal means against the alternative hypothesis of unequal distributions through unequal means.

3.2.2. Markov Chain Transition Probability Matrix. Generally, random variables X are considered as a Markov random process. For each n and all states i_1, \ldots, i_n ,

$$P[X_n = i_n | X_{n-1} = i_{n-1}, \dots, X_1 = i_1]$$

= $P[X_n = i_n | X_{n-1} = i_{n-1}].$ (4)

We use our transition matrices as in a Markovian analysis, as a consequence relative frequencies should be interpreted as probabilities; in practice we utilize the transition matrics as if they had been generated by a stationary Markov process:

$$P[X_n = j | X_{n-1} = i] = P[X_{n+k} = j | X_{n+k-1}].$$
(5)

For all states *i* and *j*, k = (n - 1), ..., 1, 0, 1, ...

3.2.3. Survival Analysis. Kaplan–Meier product limit method is used to estimate the survival function. This method is a nonparametric estimation method, which is used to estimate the survival probability beyond a given point in time; that is, the survival distribution is calculated according to the life experience data, and the censored case is considered. In other words, it is a statistical technique for describing and quantifying "time of event" data.

The survival function S(t) is estimated using the Kaplan-Meier product limit method. The specific derivation is as follows: it is assumed that the sample contains n independent observations, expressed as $(t_i; c_i)$, i = 1, 2, ..., n, where t_i is the survival time and c_i is the censored dummy variable C of the observation value i. If the "failure" event occurs, it is taken as 1; otherwise it is taken as 0. In addition, it is assumed that there are m < n recorded failures. Then, the ordered survival time $t(1) < t(2) < \cdots < t(m)$ is defined. Let n_j be the number of failures at t(j), let d_j be the number of failures observed, and the Kaplan-Meier estimate of the survival function is

$$\widehat{S}(t) = \prod_{t(i) < t} \frac{n_j - d_j}{n_j}.$$
(6)

By convention, when t < t(1), $\hat{S}(t) = 1$. Considering that many observations are censored, the estimator is robust to censoring and uses the information of censored and non-censored observations.

4. Empirical Analysis Results

4.1. Kernel Density Estimation. The kernel density of RSCA index in 2001–2002, 2008–2009, and 2019–2020 is estimated, as shown in Figure 1. The two-year average index is used to mitigate the impact of export fluctuations in a single year. At the same time, the reason for choosing the year 2001–2002 is not only the starting year of this study, but also the period of China's entry into WTO. 2008-2009 is a period of international financial crisis and 2019-2020 is the end year of the study. From the distribution pattern, there is an obvious peak on the left of the median point (RSCA = 0) in 2001-2002, indicating that most commodities have no comparative advantage. Actually, the percentages of commodity in the three years are 64.3%, 63.9%, and 62.0%, respectively, and the above results are proved. In 2008–2009, two more flat peaks are formed compared with 2001-2002. The first peak is on the left side of the peak in 2001–2002, which is comparative.

Advantage of some commodities is deteriorating; the second peak is on the right side of the peak in 2001–2002, indicating that comparative advantage of some commodities is improving. In 2019–2020, the left peak moves further to the left, indicating that comparative advantage of some commodities continues to deteriorate. On the right side of the median point (RSCA = 0), the curve in 2008–2009



FIGURE 1: Kernel density estimation of Chinese RSCA index distribution.

moves upward relative to that in 2001-2002, indicating that the number of commodities with comparative advantage increases. The curve in 2019-2020 moves further upward, indicating that the number of commodities with comparative advantage is further increased compared with the previous two years, thus forming the highest peak. Obviously, since China's accession to the WTO, although the degree of comparative disadvantage of a few export commodities has increased, it can still be seen that more and more commodities have obtained comparative advantage, and the overall trend is improving. In addition, by observing the tails at both ends of the three curves, it can be seen that the left curve moves up and the right curve moves down, which further shows that the comparative advantage of commodities with strong original advantage decreases and the comparative disadvantage of commodities with weak original disadvantage increases. This phenomenon can be explained to some extent by the calculation of RCA index. Although this study makes analysis based on RSCA index, some defects of RCA index will not be eliminated by conversion to RSCA index. Yeats believes that RCA index calculation result may be more beneficial to small economies [21]. When China joined the WTO, it was not a major exporter in the world, and only a few commodities are exported to the world market. These commodities account for a large share of Chinese total exports, while most other commodities account for a small share, which affects the numerator of RCA index $(X_{ij} / \sum_{i=1}^{m} X_{ij})$. Although Benedicit believes that the value of RCA index depends on the change of numerator and denominator $(X_{in} / \sum_{i=1}^{m} X_{in})$ and the simultaneous change of numerator and denominator [13], the author believes that the numerator influence is greater for China, thus amplifying the comparative advantage of commodities with high export share and weakening the comparative disadvantage of commodities with low export share. More than 20 years after joining WTO, China has become a major exporter. The export commodities with comparative advantage and disadvantage have a lower share compared with more than 20 years ago, which leads to decline of advantages of the commodities with strong comparative advantage and enhancement of disadvantages of the commodities with weak comparative advantage.

Wilcoxon signed rank two-tailed test is performed. The test about comparative advantage index distribution can be seen in S. Bodhisattva and D. Kaveri [22]. Here, the original hypothesis (H_0) is that there is no significant difference in the distribution of RSCA index in the above three years, while there is a significant difference in the alternative hypothesis (H_1) . The results show that the original hypothesis is not rejected at the 5% significance level, as shown in Table 1. This means that although Chinese RSCA index kernel density curve shows a certain change, the change does not deviate from the original distribution state. That is to say, despite the impact of major external environmental changes such as entry into WTO and the 2008 international financial crisis, the dynamics of Chinese export comparative advantage have changed to a certain extent, but its export specialization mode is still stable.

4.2. Transition Probability Matrix. Taking 2001–2002 as the base year, four intervals, I, II, III, and IV, are divided according to the quartile of RSCA index in 2001–2002. Interval I is between the minimum and lower quartile of RSCA index in 2001–2002, interval II is between the lower quartile and median, III is between the median and upper quartile, and IV is between the upper quartile and maximum. In this way, the transition probability matrices of 2008–2009 and 2019–2020 relative to 2001–2002 are obtained, respectively. Similarly, the other I, II, III, and IV intervals are divided based on the quartile of RSCA index in 2008–2009, and the transition probability matrix in 2019–2020 relative to 2008–2009 is also calculated. All results are shown in Table 2.

Generally speaking, the probability of the diagonal element of the matrix represents stability. When the diagonal probability value is larger, it indicates that the stability is higher. When the probability of each row of elements moves across the interval relative to probability of diagonal elements, it indicates liquidity. When crossing multiple intervals, the liquidity is greater. Since probability values of the matrix are all between [0, 1], the stability and liquidity are evaluated by summing the probability of diagonal elements and nondiagonal elements. Firstly, the transition matrix in 2019-2020 relative to 2001-2002 is analyzed. The sum of probability of diagonal elements and nondiagonal elements is 2.16 and 1.84, respectively, indicating that stability is higher than liquidity. In addition, the probability of the elements in the upper right corner outside the diagonal indicates that the liquidity is in the improving direction, and the probability of the elements in the lower left corner indicates that the liquidity is in the deterioration direction. The sum of probability of the elements in the upper right corner is 0.98, which is greater than the sum of the probability of the elements in the lower left corner which is 0.86, and the overall trend is improving. Secondly, the transition matrix in 2008-2009 relative to 2001-2002 is obtained. The sum of probability value of diagonal elements and nondiagonal

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elements is 2.5 and 1.5, respectively, and stability is higher than liquidity. Similarly, the total probability values of the upper right corner and lower left corner of the matrix are 0.73 and 0.77, respectively, and there is a deterioration trend as a whole.

Finally, the transition matrix of 2019–2020 relative to 2008–2009 is analyzed. The sum of probability value of diagonal elements is 3.09, and the sum of the probability of nondiagonal elements is 0.91; stability is higher than liquidity. The sum of the probability values in the upper right corner and the sum of the probability value of the elements in the lower left corner of the matrix are 0.52 and 0.39, respectively, showing a good trend as a whole.

From this, it can be concluded that dynamics of Chinese export comparative advantage is stable. Except that Chinese comparative advantage deteriorated slightly during international financial crisis in 2008, the overall comparative advantage has an improving trend, which also verifies the relevant results of kernel density estimation.

4.3. Survival Analysis. The above two methods only use six years' RSCA index distribution information and do not show the complete dynamics of Chinese export comparative advantage. In order to further clarify dynamics of Chinese export comparative advantage, the survival analysis of RSCA index from 2001 to 2020 is carried out. It is defined as 0 when the value of RSCA index is greater than 0 and 1 when the value is less than 0. The Kaplan–Meier method is used to estimate the cumulative survival function. Firstly, find out the uninterrupted sequence with RSCA > 0 from 2001 to 2020, which means that the value of a specific commodity in 20 years is 0. Then consider two cases.

Case 1. If RSCA > 0 turns to RSCA \leq 0 for a commodity in a certain year, it indicates that an event has occurred and is marked 1 at the end of the time sequences of successive 0's. The minimum length of the sequence is 2. The maximum length of the time sequence is 19.

Case 2. Case I does not occur. This includes two kinds of censored cases. (i) The sequence is 1 from the first year; after multiple consecutive 0's or 1's, it is finally censored with 0 in the 20th year. (ii) The sequence was 1 in 20 years and finally censored with 1.

For Kaplan Meier analysis, here, the censored case (ii) in case II is excluded, so 120 commodities are eliminated and 135 commodities remained. The following situations will happen to 135 commodities: (a) it has been 0 for 20 years; (b) Case 1 occurs; (c) there is also Case 1 and case (i) in Case 2 that occur at the same time. Thus, 173 independent observations were formed, of which 93 commodities are censored with 0, accounting for 53.7%, and 80 commodities are censored with 1, accounting for 46.3%. The survival probability of Chinese survival function in the first year is 1, which decreases to 0.661 after 5 years, 0.578 after 10 years, 0.522 after 15 years, and 0.507 after 16 years and then remains stable (see Table 3).

TABLE 1: Results for Wilcoxon's signed rank test of Chinese RSCA index.

year	2001-2002 vs. 2008-2009	2008-2009 vs. 2019-2020	2001-2002 vs. 2019-2020
<i>z</i> -value	-0.422	-0.513	-0.506
<i>p</i> -value	0.673	0.595	0.613

Note. Significance level $\alpha = 5\%$.

TABLE 2: Markov transition probability matrix of Chinese export comparative advantage.

State	2001-2002 vs. 2008-2009				2008-2009 vs. 2019-2020				2001-2002 vs. 2019-2020			
	Ι	II	III	IV	Ι	II	III	IV	Ι	II	III	IV
Ι	0.734	0.219	0.031	0.016	0.891	0.109	0.000	0.000	0.672	0.219	0.094	0.015
II	0.210	0.500	0.290	0.000	0.143	0.603	0.254	0.000	0.238	0.302	0.429	0.031
III	0.063	0.187	0.578	0.172	0.000	0.110	0.734	0.156	0.063	0.219	0.531	0.187
IV	0.046	0.077	0.185	0.692	0.000	0.000	0.141	0.859	0.094	0.047	0.203	0.656

TABLE 3: Kaplan-Meier survival analysis of Chinese RSCA index (2001-2020).

Time	Beg. total	Fail	Net lost	Survivor function	Std. error	[95% co	[95% conf. int.]	
1	173	0	2	1.0000				
2	171	23	2	0.8655	0.0261	0.8046	0.9085	
3	146	14	2	0.7825	0.0316	0.7126	0.8373	
4	130	12	3	0.7103	0.0349	0.6355	0.7725	
5	115	8	4	0.6609	0.0366	0.5837	0.7271	
6	103	2	0	0.6480	0.0370	0.5704	0.7152	
7	101	4	0	0.6224	0.0377	0.5438	0.6913	
8	97	3	0	0.6031	0.0381	0.5241	0.6732	
9	94	3	0	0.5839	0.0385	0.5045	0.6550	
10	91	1	2	0.5775	0.0386	0.4980	0.6489	
11	88	2	2	0.5643	0.0388	0.4847	0.6364	
13	84	0	4	0.5643	0.0388	0.4847	0.6364	
14	80	3	0	0.5432	0.0392	0.4631	0.6163	
15	77	3	4	0.5220	0.0396	0.4417	0.5962	
16	70	2	3	0.5071	0.0398	0.4266	0.5820	
17	65	0	1	0.5071	0.0398	0.4266	0.5820	
18	64	0	1	0.5071	0.0398	0.4266	0.5820	
19	63	0	2	0.5071	0.0398	0.4266	0.5820	
20	61	0	61	0.5071	0.0398	0.4266	0.5820	



FIGURE 2: Kaplan-Meier survival analysis of Chinese RSCA index (2001-2020).

Graphically, the period of rapid decline in Chinese survival rate is mainly 1–5 years, the degree of decline decreases in 5–15 years and remains stable after 16 years (see Figure 2). This means that about 50% of China's export commodities with comparative advantage have a chance to survive for more than 16 years. At the 95% confidence level, the mean survival time is 12.84 years and the standard error is 0.61. This shows that, even in the face of fierce international market competition, after excluding the commodities without comparative advantage in the past 20 years, the viability of China's commodities with comparative advantage performs well, which means that China's existing export mode can still support the viability of most commodities.

5. Conclusions and Suggestions

Since entrance to WTO, the development of Chinese foreign trade has significantly improved its position in

world trade and become the largest commodity trade country and largest exporter. This is the result of China's continuous optimization of import and export trade mode to meet the needs of world market based on its own comparative advantages. Based on the RSCA index, this paper studies the dynamics of Chinese export comparative advantage from 2001 to 2020; by kernel density estimation method, it is obtained that although China is facing the impact of more fierce international market competition after entrance to WTO, the export comparative advantage of a small number of commodities has weakened, and comparative disadvantage has increased, but most of commodities' comparative advantage has improved, and through Wilcoxon signed rank two-tailed test, it is concluded that the original comparative advantage state of China has not changed. Through analysis of Markov chain transition probability matrix, it is concluded that Chinese export trade mode is relatively stable, the stability of comparative advantage is higher than liquidity, and the liquidity presents an improving trend as a whole. From survival analysis, after excluding the commodities that have not had comparative advantages for 20 years, the viability of China's commodities with comparative advantages performs well, which means that existing export mode can still support viability of most commodities.

As COVID-19 continues to rage, competition between China and major trading partners in trade and other fields will have a greater impact on world trade. Therefore, China needs to be based on the reality and evolution characteristics of export commodities comparative advantage, adapt to the dynamic demand change of international market, actively optimize specialized export mode, and enhance comparative advantage. Specifically, firstly, China should further clarify the status and trend of comparative advantage of various commodities in world market and major export markets, adapt to market dynamic demand, strengthen product innovation, improve the supply chain, improve level of value chain, and further improve added value of export commodities so as to stabilize and develop comparative advantage. Secondly, China should focus on the export of commodities with improved comparative advantages, promote diversification of export markets, and further improve the way of trade organization so as to promote the release of potential of such commodities. Thirdly, China should continue to strengthen the existing export mode and improve the viability of export products continuously in international market based on comparative advantage.

Data Availability

The data used to support the findings of this study are available from the corresponding author upon request.

Conflicts of Interest

The authors declare no conflicts of interest regarding the content and implications of this manuscript.

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Research Article

Pseudospectral Method Based on Müntz–Legendre Wavelets for Solving the Abel Integral Equation

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This paper deals with the numerical solution of the Abel integral equation based on Müntz–Legendre wavelets. To this end, the Abel integral operator is represented by Müntz–Legendre wavelets as an operational matrix. To find this matrix, we use the similarity between the Abel integral operator and the fractional integral operator. The proposed method can be easily used to solve weakly singular Volterra integral equations. We have proved the convergence of the proposed method. To demonstrate the ability and accuracy of the method, some numerical examples are presented.

1. Introduction

In this paper, we focus our attention on constructing and applying Müntz–Legendre (M-L) wavelets that will be used as the basis in the pseudospectral method to solve the famous Abel integral equation

$$u(x) - \mathscr{A}^{\alpha}(k)(x) = f(x), \qquad (1)$$

in which Abel's integral operator $\mathscr{A}^{\alpha}(k)(x)$ of order $0 < \alpha < 1$ is defined in [1] as follows:

$$\mathscr{A}^{\alpha}(k)(x) \coloneqq \int_{0}^{x} k(x,s,u(s))(x-s)^{-\alpha} \mathrm{d}x, \quad x \in [0,1].$$
(2)

Here, given $\Omega = [0, 1]$, f(x), and k(x, s, u(s)) are assumed to be continuous functions on Ω and S with $S = \{(x, s, u): x, s \in [0, 1], u \in \mathbb{R}\}$. Further, we suppose that the kernel function k(x, s, u(s)) is equal to the form g(x, s)u(s). In other words, the desired equation is assumed to be linear.

The Abel equation is a special case of the integral equations with the weakly singular kernel that was first introduced by Abel. In investigating the generalization of the tautochrone problem, he introduced this equation [2]. This equation appears widely in modeling many physical problems, such as nuclear physics, X-ray radiography, fluid flow [3], scattering theory, plasma diagnostics, semiconductors, physical electronics, and nonlinear diffusion[1, 4]. Given this equation's wide application, solving this equation is very important. But one cannot always solve the equation analytically, and we need to use numerical methods for it.

Among the many papers that have considered the numerical solution of this equation, we can mention some of them. Saadatmandi and Dehghan [5] utilized the collocation method based on shifted Legendre polynomials. Piessens and Verbaeten [6] introduced a numerical method based on Chebyshev polynomials, and after approximating the unknown solution based on these bases, they obtained the solution as a sum of hypergeometric functions. Using the Bernstein operational matrix, Singh et al. [7] introduced a stable numerical method to solve this problem. In [8], we can find the integrable solution of the Abel integral equation under certain conditions, and also the sufficient and necessary conditions for the existence of this solution are presented. In [9], the authors proposed the Laplace transform method to solve the problem, where they assumed that the solution would be differentiable and continuous. Saray [10] introduced a novel and efficient method based on

Alpert's multiwavelets. In this work, after introducing the sparse representation of the Abel integral operator, the Abel integral equation is reduced to a sparse system of linear algebraic equations in the linear form, and this causes a reduction in time and computational costs. In [11], the unbounded solutions of the nonlinear Abel integral equations are investigated. Li and Zhao [12] used Mikusinski's operator of fractional order to solve the Abel integral equation.

The outline of this article is as follows: the M-L wavelets are constructed in Section 2, and then the Abel integral operator is represented based on these bases. In Section 3, the Abel integral equation is solved by using the pseudospectral method based on M-L wavelets. This section contains the error analysis and the conditions for convergence are investigated. To demonstrate the efficiency and accuracy of the presented method, some numerical examples are given in Section 4.

2. Müntz-Legendre Wavelets

As we know, multiresolution analysis (MRA) is a significant procedure for constructing wavelets. According to MRA, a family of nested subspaces exists such that they satisfy certain circumstances [13]

$$\{0\} \subset \cdots \subset V_{-1} \subset V_0 \subset V_1 \subset \cdots \subset L^2(\Lambda), \tag{3}$$

where Λ is equal to \mathbb{R} or any bounded interval.

Recently, the M-L wavelets have been used to solve some equations, such as fractional optimal control problems [14], fractional pantograph differential equations [15], fractional differential equations [16], and multiorder fractional differential equations [17]. To solve the Abel equation, we first briefly introduce the M-L wavelets as follows.

Given $J \in \mathbb{N}_0$, assume that the subspace $V_J \in L^2(\Lambda)$ is spanned by the scaled and translated version of a set of bases, which are called multiscaling functions, viz.,

$$V_{J} = \left\{ \operatorname{span} \left\{ \phi_{J,b}^{n} \colon b \in \mathscr{B}, n \in \mathscr{R} \right\} \right\},$$
(4)

where $\mathscr{B} \coloneqq \{0, 1, ..., 2^J - 1\}$ and $\mathscr{R} \coloneqq \{0, 1, ..., r - 1\}$ with $r \in \mathbb{N}$. The parameter *J* is called refinement level and *r* is the multiplicity parameter. In the sequel, we intend to introduce the functions $\phi_{J,b}^n$.

Motivated by [17], we denote the M-L polynomials $L_n(x)$ as

$$L_n(x) = \sum_{k=0}^n l_{k,n} x^{\lambda_k}, \quad x \in \Omega,$$
(5)

where $\lambda_k := \{k\mu: \mu \in \mathbb{R}, k = 0, ..., n\}$ and the coefficient $l_{k,n}$ is obtained by

$$l_{k,n} \coloneqq \frac{\prod_{i=0}^{n-1} \left(\lambda_k + \lambda_i + 1\right)}{\prod_{i=0, i \neq k}^n \left(\lambda_k - \lambda_i\right)}.$$
(6)

It can be easily shown that these polynomials satisfy the orthogonality requirements and form an orthogonal system, via

$$\langle L_n(x), L_{n'}(x) \rangle = \int_0^1 L_n(x) L_{n'}(x) dx = \frac{\delta_{n',n}}{2\lambda_n + 1}, \quad n \ge n',$$
(7)

where $\delta_{m,m}$ denotes the Kronecker symbol and is given by

$$\delta_{n',n} \coloneqq \begin{cases} 1, & n' = n, \\ 0, & n \ge n'. \end{cases}$$
(8)

Considering the definition of $L_n(x)$, one can introduce the M-L wavelets [17], via

$$\phi_{J,b}^{n} = \begin{cases} 2^{J/2} \sqrt{2\lambda_{n} + 1} L_{n} (2^{J} x - b), & \frac{b}{2^{J}} \le x \le \frac{b+1}{2^{J}}, \\ 0, & \text{otherwise.} \end{cases}$$
(9)

Due to the definition of M-L wavelets, one can introduce the projection operator \mathscr{P} that maps any function $u \in L^2(\Omega)$ onto V_I as follows:

$$u(x) \approx \mathscr{P}(u)(x) = \sum_{b=0}^{2^{J-1}} \sum_{n=0}^{r-1} u_{b,n} \phi_{J,b}^{n}(x) = U^{T} \Phi(x), \quad (10)$$

where and throughout the paper, the superscript *T* is used for the matrix transpose. Here $\Phi(x)$ is a vector function of dimension $N = 2^{J}r$ whose (br + n + 1)-th element is $\phi_{J,b}^{n}(x)$, and the (br + n + 1)-th element of the vector *U* is evaluated by

$$u_{b,n} = \langle u, \phi_{J,b}^n \rangle = \int_0^1 u(x) \phi_{J,b}^n(x) \mathrm{d}x.$$
(11)

It follows from [15] that one may be able to bound the projection error \mathcal{P} in the sense of Sobolev norms.

Lemma 1 (see [15]). Given $n \ge 0$, assume that r > m. If $u \in H^m(\Omega)$, then

$$\|u - \mathscr{P}(u)\|_{L_{2}(\Omega)} \le c (r-1)^{-m} (2^{J-1})^{-m} \|u^{(m)}\|_{L_{2}(\Omega)}, \quad (12)$$

and for $s \ge 1$, we have

$$\|u - \mathscr{P}(u)\|_{H^{s}(\Omega)} \le c (r-1)^{2s-(1/2)-m} (2^{J-1})^{s-m} \|u^{(m)}\|_{L_{2}(\Omega)},$$
(13)

in which $H^m(\Omega)$ is the Sobolev space and the related norm is determined by

$$\|u\|_{H^{m}(\Omega)} = \left(\sum_{j=0}^{m} \|u^{(j)}\|_{L_{2}(\Omega)}^{2}\right)^{1/2}.$$
 (14)

2.1. Representation of Abel Integral Operator in Müntz-Legendre Wavelets. In this subsection, we consider the Abel operator as a fractional integral operator, and after representing the fractional integral operator in M-L wavelets as an operational matrix, we find a representation of the Abel

integral operator in M-L wavelets. To this end, it is necessary to define some concepts about fraction calculation.

Definition 1 (see [18]). Let $u \in L_1[a, b]$. The Riemann-Liouville fractional integral operator \mathscr{F}_a^{α} of order $\alpha \in \mathbb{R}^+$ is determined by

$$\mathscr{I}_{a}^{\alpha}(u)(x) \coloneqq \frac{1}{\Gamma(\alpha)} \int_{a}^{x} (x-s)^{\alpha-1} u(s) \mathrm{d}s, \quad x \in [a,b], \quad (15)$$

where $\Gamma(\alpha)$ represents the gamma function.

Remark 1. It can be verified that the fractional integral of the functions x^{κ} is given by

$$\mathcal{F}_{a}^{\alpha}(x^{\kappa}) = \frac{\Gamma(\kappa+1)}{\Gamma(\kappa+\alpha+1)} x^{\kappa+\alpha}.$$
 (16)

Lemma 2 (cf Lemma 2.1(a). see [19]). The fractional integration operators \mathcal{F}_a^{α} are bound in $L^p([a, b])$ for $1 \le p \le \infty$ as follows:

$$\left\|\mathscr{F}_{a}^{\alpha}(u)\right\|_{p} \leq K \|u\|_{p}, \quad K \coloneqq \frac{(b-a)^{\alpha}}{\Gamma(\alpha+1)}.$$
(17)

It follows from [19] that if $u \in L_1[a, b]$, then the function $\mathscr{F}_a^{\alpha} u$ itself belongs to $L_1[a, b]$. We recall that the Abel integral operator of order $\alpha \in (0, 1)$ is determined by

$$\mathscr{A}^{\alpha}(u)(x) = \int_0^x (x-s)^{-\alpha} u(s) \mathrm{d}s.$$
 (18)

There is a similarity between the Abel integral operator \mathscr{A}^{α} and the Riemann–Liouville fractional integral operator \mathscr{F}^{β}_{0} ($\beta \coloneqq 1 - \alpha$), viz,

$$\mathscr{A}^{\alpha} = \Gamma(\alpha)\mathscr{F}_{a}^{1-\alpha}.$$
 (19)

Thus, we can use the Riemann–Liouville fractional integral operator \mathscr{F}_0^{β} instead of the Abel integral operator \mathscr{A}^{α} .

According to the definition of M-L wavelets, the fractional integral operator \mathcal{F}_0^β acting on the vector function $\Phi(x)$ may be written as an expansion of M-L wavelets $\Phi(x)$, i.e.,

$$\mathscr{P}\left(\mathscr{F}_{0}^{\beta}\right)(\Phi(x)) = I_{\beta}(x)\Phi(x), \quad \beta \in (0,1),$$
(20)

where $I_{\beta}(x)$ is an $N \times N$ matrix and is famous as the operational matrix of fractional integral for the M-L wavelets.

Before we look at how to calculate the elements of the aforementioned matrix, it is intransitive to introduce the piecewise fractional-order Taylor functions. Let $J \in \mathbb{Z}^+ \cup \{0\}$ be a fixed number, the piecewise fractional-order Taylor functions can be defined in the following form:

$$\psi_{J,b}^{n} = \begin{cases} t^{\lambda_{n}}, & \frac{b}{2^{J}} \le x \le \frac{b+1}{2^{J}}, \\ 0, & \text{otherwise.} \end{cases} \quad b \in \mathcal{B}, n \in \mathcal{M}.$$
(21)

By introducing the square matrix T of dimension $N \times N$, whose (i, j)-th element is computed by

$$T_{i,j} = \langle \Phi_i(x), \Psi_j(x) \rangle = \int_0^1 \Phi_i(x), \Psi_j(x) dx,$$

$$i, j = 1, \dots, N.$$
 (22)

We can expand any elements of vector function $\Phi(x)$ (M-L wavelets) by the piecewise fractional-order Taylor functions $\Psi(x)$, viz,

$$\Phi(x) = T^{-1}\Psi(x). \tag{23}$$

Here the matrix *T* is a $N \times N$ matrix and it is called the transformation matrix. In the sequel, we assume that *Q* is a vector of dimension *r* whose *i*-th element is x^{λ_i} . Thus, it is easy to show that

$$\Psi(x) = [Q, \dots, Q]^T.$$
⁽²⁴⁾

It follows from (16) that one can find the *i*-th element of $\mathscr{F}_0^{\beta}(\Psi)(x)$, via

$$\mathscr{I}_{0}^{\beta}(\Psi_{i})(x) = \frac{\Gamma(\lambda_{i}+1)}{\Gamma(\lambda_{i}+\beta+1)} x^{\lambda_{i}+\beta}.$$
 (25)

This gives rise to introduce a diagonal matrix $I_{\Phi,\beta}(x)$, such that

$$\mathscr{I}_{0}^{\beta}(\Psi)(x) = I_{\Psi,\beta}(x)\Psi(x). \tag{26}$$

It is worth noting that this matrix expresses the fractional integral of functions Ψ as a combination of themselves and has the following form:

$$I_{\Psi,\beta}(x) = \operatorname{diag} \left[P_{\beta}(x), \dots, P_{\beta}(x) \right],$$
(27)

where $P_{\beta}(x) \coloneqq x^{\beta}H(\mathscr{F}_{0}^{\beta}(Q)(x) = P_{\beta}(x)Q(x))$ and H is a diagonal matrix of the form

$$(H)_{i,i} = \frac{\Gamma(\lambda_i + 1)}{\Gamma(\lambda_i + \beta + 1)}.$$
(28)

Now, we are able to introduce the operational matrix of fractional integral for the M-L wavelets via

$$\mathcal{P}_{J}(\mathcal{F}_{0}^{\beta})(\Phi(x)) = \mathcal{P}_{J}(\mathcal{F}_{0}^{\beta})(T^{-1}\Psi(x))$$
$$= T^{-1}I_{\Psi,\beta}(x)\Psi(x) = T^{-1}I_{\Psi,\beta}(x)T\Phi(x).$$
(29)

Thus, we get

$$I_{\beta}(x) \coloneqq T^{-1}I_{\Psi,\beta}(x)T.$$
(30)

3. Pseudospectral Method

To derive the numerical solution of the second kind of Abel integral equation based on the pseudospectral method, we can approximate the unknown solution with the projection operator \mathcal{P}_I , as follows:

$$u \approx \mathscr{P}_{I}(u) = U^{T} \Psi, \qquad (31)$$

where U is a vector of dimension N, whose elements should be found. Note that the function f(x) and the kernel function $k(x, s, \mathcal{P}_{J}(u)(s))$ can be approximated in the same manner, i.e.,

$$f(x) \approx \mathscr{P}_{J}(u) = F^{T}\Psi,$$

$$k(x, s, \mathscr{P}_{J}(u)(s)) \approx \mathscr{P}_{J}(k)(x, s, \mathscr{P}_{J}(u)(s))$$
(32)

$$= \Psi^{T}(x)K\Psi(s),$$

where $F \in \mathbb{R}^N$ and $K \in \mathbb{R}^{N \times N}$. By substituting (32) into the integral part of the Abel integral equation (1), we obtain

$$\mathcal{P}_{J}\left(\Psi(x)^{T}\widetilde{K}\int_{0}^{x}(x-s)^{-\alpha}\Psi(s)\mathrm{d}s\right) = \mathcal{P}_{J}^{r}\left(\Psi(x)^{T}\widetilde{K}I_{\alpha}\Psi(x)\right)$$
$$= \widetilde{K}\Psi(x).$$
(33)

We now substitute equations (31)–(33) into Abel integral equation (1) and simplify to get

$$r(x) \coloneqq (U - F + \widehat{K})^T \Psi(x) = 0, \qquad (34)$$

where r(x) is the residual function that our goal is to reduce to zero. Let $\{x_i\}$ be a number of points in Ω , we select the solution that satisfies the collocation condition $r(x_i) = 0$, where $\{x_i\}$ are called the collocation points. In this paper, we use the shifted Chebyshev and Legendre polynomials zeros as collocation points. The collocation method gives rise to a system of linear or nonlinear algebraic equations. One can derive the unknown coefficients U after solving this system.

3.1. Error Analysis. We write the Abel integral equation (1) in the form

$$(I - \mathscr{K})u = f, \tag{35}$$

where \mathcal{K} is a compact operator that maps any continuous function onto C[0, 1]. As we said, our goal is to reduce the residual function r(x) to zero. Symbolically, we have

$$r(x) = (I - \mathcal{K})u_I - f, \qquad (36)$$

where $u_j := \mathscr{P}_j(u)$. We note that $\mathscr{P}_j(r)(x) = 0$ if and only if $r(x_i) = 0$ or equivalently,

$$\mathscr{P}_{I}(I-\mathscr{K})u_{I}=\mathscr{P}_{I}(f). \tag{37}$$

Let u_I is a solution of (37), then by applying $\mathscr{P}_I(u_I) = u_I$, equation (37) can be written as follows:

$$(I - \mathscr{P}_J \mathscr{K}) u_J = \mathscr{P}_J(f).$$
 (38)

Since both the original equations (1) and (38) are defined on C[0, 1], then for the error analysis, we compare them.

Theorem 1. Let us assume that $I - \mathscr{K}: C[0,1] \longrightarrow C[0,1]$ is a bijections operator. Further, let us assume

$$\|\mathscr{K} - \mathscr{P}_J \mathscr{K}\| \longrightarrow 0, \quad \text{as } r \longrightarrow \infty.$$
 (39)

Then for all sufficiently large r $(r \le N)$, the operator $(I - \mathcal{P}_I \mathcal{K})^{-1}$ exists as a bounded operator from C[0, 1] to C[0, 1]. Moreover, it is uniformly bounded

$$\sup_{r \le N} \left\| \left(I - \mathscr{P}_J \mathscr{K} \right)^{-1} \right\| < \infty.$$
(40)

For the solutions of (35) and (38)

$$u - u_{J} = \left(I - \mathscr{P}_{J}\mathscr{K}\right)^{-1} \left(u - \mathscr{P}_{J}(u)\right),$$

$$\frac{1}{\left\|I - \mathscr{P}_{J}\mathscr{K}\right\|} \left\|u - \mathscr{P}_{J}(u)\right\| \leq \left\|u - u_{J}\right\|$$

$$\leq \left\|\left(I - \mathscr{P}_{J}\mathscr{K}\right)^{-1}\right\| \left\|u - \mathscr{P}_{J}(u)\right\|.$$
(41)

Proof. For details, refer to [20].

Since $\mathscr{K}: C[0,1] \longrightarrow C[0,1]$ is a compact operator and \mathscr{P}_J is a bounded projection, such that $\mathscr{P}_J u \longrightarrow 0$ as $r \longrightarrow \infty$. Then motivated by [20] (Lemma 3.1.2), we have

$$\|\mathscr{K} - \mathscr{P}_{J}\mathscr{K}\| \longrightarrow 0, \quad \text{as } r \longrightarrow \infty.$$
 (42)

Therefore, the condition of Theorem 1 is held. \Box

4. Numerical Examples

In this section, some numerical examples are solved to show the validity and efficiency of the method. To do this, we carry out the Maple and MATLAB software simultaneously.

Example 1. For the first example, let us consider the linear Abel integral equation of the second kind with the kernel function $k(t, x, u(x)) \coloneqq (1/10\Gamma(1 - \alpha))u(x)$, and f(t) = 1. The exact solution is given in [10] as follows:

$$u(t) = E_{1-\alpha,1}\left(\frac{t^{1-\alpha}}{10}\right),$$
(43)

where $E_{\sigma,\beta}$ is the Mittag-Leffler function

$$E_{\sigma,\beta}(z) = \sum_{l=0}^{\infty} \frac{z^l}{\Gamma(\sigma l + \beta)}, \quad \sigma, \beta, z \in \mathbb{R}, \ \sigma > 0.$$
(44)

Table 1 shows the absolute value of errors at different times x_i when the collocation points are chosen to be the Legendre polynomial nodes. As we expected, when the r increases (the degree of the bases as well as the number of collocation points increases) the error will decrease. To show the effect of the parameter μ in the L^2 -error, we plot Figure 1 and report Table 2. Figure 2 illustrates the effect of the choosing nodes on the L^2 -error and also absolute value of errors at the Chebyshev nodes using different multiplicity r. Also, we can see the effect of the multiplicity parameter r on the L^2 -error and absolute error in Figure 2.

Example 2. The second example is devoted to the Abel integral equation (1) with

$$f(t) \coloneqq 2\sqrt{t},$$

$$k(t, x, u(x)) \coloneqq u(x).$$
(45)

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<u>x</u>	<i>r</i> = 5	<i>r</i> = 7	<i>r</i> = 9
0.1	9.72e - 4	2.78e - 4	1.19e – 4
0.2	2.41e - 4	2.24e - 4	1.30e - 5
0.3	3.15e - 4	1.22e - 6	4.64e - 5
0.4	2.93e - 4	1.09e - 4	4.83e - 5
0.5	1.23e - 5	7.38 <i>e</i> – 7	1.93e – 6
0.6	1.94e - 4	7.95e - 6	3.14e - 5
0.7	1.53e - 4	6.80e - 6	2.21 <i>e</i> – 5
0.8	7.94e - 5	6.32e - 5	6.04e - 6
0.9	1.89e - 4	4.44e - 5	1.60 <i>e</i> – 5
1.0	4.09e - 4	1.60e - 4	7.86 <i>e</i> – 5

TABLE 1: The absolute errors at the Legendre nodes for Example 1.



FIGURE 1: Plot of the L^2 -errors taking $\mu = (1/2)$ (a) and $\mu = (1/3)$ (b) at the Legendre and Chebyshev nodes for Example 1.

TABLE 2: The L^2 -errors at the Legendre and Chebyshev nodes taking different μ for Example 1. $\mu = (1/2)$ $\mu = (1/3)$

r	$\mu = (1/2)$		$\mu = (1/3)$	
	Chebyshev nodes	Legendre nodes	Chebyshev nodes	Legendre nodes
2	7.57e - 4	7.87e - 4	4.03e - 3	4.48e - 3
3	1.69 <i>e</i> – 5	2.06e - 5	9.34 <i>e</i> – 5	9.79e – 5
4	3.82e - 7	5.32e – 7	9.81 <i>e</i> – 5	1.27e - 4
5	8.58e – 9	1.33 <i>e</i> – 8	3.37 <i>e</i> – 5	4.29e - 5
6	1.89e - 10	3.16 <i>e</i> – 10	1.76 <i>e</i> – 5	2.26e - 5
7	4.06e - 12	7.19 <i>e</i> – 12	9.89 <i>e</i> – 6	1.27 <i>e</i> – 5

The exact solution is given by $u(t) = 1 - e^{\pi t} erfc(\sqrt{\pi t})$ [4].

To show the effect of the multiplicity parameter r and choosing the collocation points, we report Table 3. This Table also illustrates the effect of parameter μ on the L^2 -error. Due to Table 3, it is obvious that these three parameters have a direct effect on the L^2 -error such that when r increases, the error decreases. Also, choosing the Chebyshev nodes gives us a better result than Legendre nodes. In Figure 3, we

demonstrate the absolute error when multiplicity parameter r increases taking $\mu = (1/2)$ and different collocation points.

Example 3. Let us consider the following Abel integral equation:

$$f(t) \coloneqq \sin(x) - \frac{4}{3}x^{3/2} {}_{1}F_{2}\left(1; \frac{5}{4}, \frac{7}{4}; -\frac{x^{2}}{4}\right), \quad k(t, x, u(x)) \coloneqq u(x),$$
(46)



FIGURE 2: Plot of the L^2 -errors (a) taking the Legendre and Chebyshev nodes and the absolute error at the Chebyshev nodes (b) for Example 1.

		0 1	0 1 1	
r	$\mu = (1/2)$		$\mu = 1$	
	Chebyshev nodes	Legendre nodes	Chebyshev nodes	Legendre nodes
2	3.38 <i>e</i> – 2	3.52e - 2	5.68 <i>e</i> – 2	5.62e - 2
3	9.21e - 2	1.08e - 2	3.15 <i>e</i> – 2	3.15 <i>e</i> – 2
4	2.57 <i>e</i> – 3	3.36 <i>e</i> – 3	1.95e - 2	2.00e - 2
5	7.29e - 4	1.04e - 3	1.31 <i>e</i> – 2	1.38e - 2
6	2.06e - 4	3.17e - 4	9.40e - 3	1.01e - 2
7	5.77 <i>e</i> – 5	9.40 <i>e</i> – 5	7.04e - 3	7.73 <i>e</i> – 3

TABLE 3: The L^2 -errors at the Legendre and Chebyshev nodes taking different μ for Example 2.



FIGURE 3: Plot of the absolute error at the Chebyshev nodes (a) and the Legendre nodes (b) taking different r for Example 2.

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x	<i>r</i> = 6	r = 8	<i>r</i> = 10
0.1	1.42e - 4	1.20e - 6	1.60 <i>e</i> – 7
0.2	6.05e - 6	2.45e - 6	1.53e – 9
0.3	2.39e - 6	3.44e - 7	4.50e - 8
0.4	4.67e - 5	6.55 <i>e</i> – 7	6.19 <i>e</i> – 8
0.5	8.29 <i>e</i> – 5	1.61 <i>e</i> – 6	7.59 <i>e</i> – 8
0.6	1.03e - 4	1.87e - 6	1.08e - 7
0.7	1.23e - 4	2.31e - 6	1.48e - 7
0.8	1.66e - 4	3.36 <i>e</i> – 6	2.01e - 7
0.9	2.40e - 4	4.64e - 6	2.76 <i>e</i> – 7
1.0	3.21 <i>e</i> – 4	6.34 <i>e</i> - 6	3.78 <i>e</i> – 7

TABLE 4: The absolute errors at the Chebyshev nodes for Example 3.



FIGURE 4: Plot of the L^2 -errors taking $\mu = (1/2)$ at the Legendre and Chebyshev nodes for Example 3.

in which ${}_{1}F_{2}$ is the hypergeometric function defined in [21]. Also, the exact solution is $u(x) = \sin(x)$.

Table 4 shows the absolute value of errors at different times x_i when the collocation points are chosen to be the Chebyshev polynomials nodes. Figure 4 illustrates the effect of selecting the Chebyshev and Legendre nodes. We can also see the effect of increasing the parameter r. It is observed that by increasing the parameter r, the error decreases.

5. Conclusion

In this paper, we utilize an efficient algorithm based on the wavelet pseudospectral method to solve the well-known Abel integral equation. This method can easily be used to solve weakly singular Volterra integral equations, and this shows the ability of the proposed method. We have compared the method with other methods and shown that this method offers better results. We have proved the convergence of the proposed method. Given the construction of these bases and the role of the parameter μ , which can be

polynomials with fractional powers, compared to other bases, if the exact solution or the known functions in the equation are of the fractional type, the proposed method will provide better results.

Data Availability

The data used to support this study are included within this article.

Conflicts of Interest

The authors declare that there are no conflicts of interest with this work.

Authors' Contributions

All authors read and approved the final manuscript. All authors contributed equally and significantly to the writing of this paper.

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Research Article

A Comparative Analysis of the Fractional-Order Coupled Korteweg–De Vries Equations with the Mittag–Leffler Law

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This article applies efficient methods, namely, modified decomposition method and new iterative transformation method, to analyze a nonlinear system of Korteweg–de Vries equations with the Atangana–Baleanu fractional derivative. The nonlinear fractional coupled systems investigated in this current analysis are the system of Korteweg–de Vries and the modified system of Korteweg–de Vries equations applied as a model in nonlinear physical phenomena arising in chemistry, biology, physics, and applied sciences. Approximate analytical results are represented in the form of a series with straightforward components, and some aspects showed an appropriate dependence on the values of the fractional-order derivatives. The convergence and uniqueness analysis is carried out. To comprehend the analytical procedure of both methods, three test examples are provided for the analytical results of the time-fractional KdV equation. Additionally, the efficiency of the mentioned procedures and the reduction in calculations provide broader applicability. It is also illustrated that the findings of the current methodology are in close harmony with the exact solutions. The series result achieved applying this technique is proved to be accurate and reliable with minimal calculations. The numerical simulations for obtained solutions are discussed for different values of the fractional order.

1. Introduction

Many researchers have been working on various aspects of fractional derivatives in recent years. Caputo and Fabrizio modified the existing Caputo derivative to develop the Caputo–Fabrizio fractional derivative [1–5] based on a nonsingular kernel. Because of its advantages, numerous researchers utilized this operator to investigate various types of fractional-order partial differential equations [6–9]. To address this issue, Atangana and Baleanu proposed a new fractional operator called the Atangana–Baleanu derivative, which combines Caputo and Riemann–Liouville derivatives. Because of the existence of the Mittag–Leffler kernel, which is a generalization of the exponential kernel, this new

Atangana–Baleanu derivative has a long memory. Moreover, the Atangana–Baleanu operator outperforms other operators, and different scientific models have been successfully solved. Many advances have been made in fractional calculus over the last few years by borrowing ideas from classical calculus, but it does not remain easy. Scholars have the main concern to obtain a numerical solution; for this, numerous efficient methodologies have been constructed for fractional differential equations, such as the Adomian decomposition transform method [10], variational iteration transform method [11, 12], optimal homotopy asymptotic method [13], and homotopy perturbation method [14, 15].

Korteweg and de Vries introduced the Korteweg-de Vries equation in 1895 to model shallow water waves in a

canal [16]. The suggested system Korteweg-de Vries equations play a crucial role in diverse engineering and applied sciences such as plasma physics, water waves, hydrodynamics, and theory of the quantum field. The Korteweg-de Vries equations are usually investigated in the analysis of nonlinear dispersive waves [17]. They define the interactions among two long waves with various dispersion relations. Many researchers have been interested in these schemes, and a lot of works have been done. For example, Ghoreishi et al. applied the homotopy analysis method to achieve numerical results of a modified system of Korteweg-de Vries equations [18]. Kaya and Inan in [19] achieved traveling wave results of the system of Korteweg-de Vries and modified system of Korteweg-de Vries equations. The fractional-order system of Korteweg-de Vries equations is defined as follows:

$$\frac{\partial^{\gamma} \mathbb{U}}{\partial \mathfrak{F}^{\gamma}} = -\rho \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - 6\rho \mathbb{U} \frac{\partial \mathbb{U}}{\partial \varphi} + 6\mathbb{V} \frac{\partial \mathbb{V}}{\partial \varphi},$$
(1)
$$\frac{\partial^{\gamma} \mathbb{V}}{\partial \mathfrak{F}^{\gamma}} = -\rho \frac{\partial^{3} \mathbb{V}}{\partial \varphi^{3}} - 39\mathbb{U} \frac{\partial \mathbb{V}}{\partial \varphi}, \quad \mathfrak{F} > 0, 0 < \gamma \le 1,$$

where γ is the fractional-order derivative of $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$, ϑ , and ρ are constants, respectively. The functions $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$ are considered as important functions of time and space, disappearing for \mathfrak{F} and φ , respectively. The other method eliminates to the conventional coupled Korteweg–de Vries equations since $\rho = \vartheta = 1$ is implemented.

A classic model in this hierarchy is the modified coupled Korteweg–de Vries system. The following nonlinear partial differential equations govern this model [20]:

$$\begin{split} \frac{\partial^{\gamma} \mathbb{U}}{\partial \mathfrak{F}^{\gamma}} &= \frac{1}{2} \frac{\partial^{3} \mathbb{U}}{\partial \mathfrak{F}^{3}} - 3 \mathbb{U}^{2} \frac{\partial \mathbb{U}}{\partial \varphi} + \frac{3}{2} \mathbb{W} \frac{\partial^{2} \mathbb{V}}{\partial \varphi^{2}} + 3 \frac{\partial \mathbb{V}}{\partial \varphi} \frac{\partial \mathbb{W}}{\partial \varphi} \\ &\quad + \frac{3}{2} \mathbb{V} \frac{\partial^{2} \mathbb{W}}{\partial \varphi^{2}} + 3yx \frac{\partial \mathbb{U}}{\partial \varphi} + 3zx \frac{\partial \mathbb{V}}{\partial \varphi} + 3zy \frac{\partial \mathbb{W}}{\partial \varphi}, \\ \frac{\partial^{\gamma} \mathbb{V}}{\partial \mathfrak{F}^{\gamma}} &= -\frac{\partial^{3} \mathbb{V}}{\partial \varphi^{3}} - 3 \frac{\partial \mathbb{U}}{\partial \varphi} \frac{\partial \mathbb{V}}{\partial \varphi} - 3 \mathbb{V} \frac{\partial^{2} \mathbb{U}}{\partial \varphi^{2}} - 3 \mathbb{V}^{2} \frac{\partial \mathbb{W}}{\partial \varphi} \\ &\quad + 6zy \frac{\partial \mathbb{U}}{\partial \varphi} + 3 \mathbb{U}^{2} \frac{\partial \mathbb{V}}{\partial \varphi}, \\ \frac{\partial^{\gamma} \mathbb{W}}{\partial \mathfrak{F}^{\gamma}} &= -\frac{\partial^{3} \mathbb{W}}{\partial \varphi^{3}} - 3 \frac{\partial \mathbb{U}}{\partial \varphi} \frac{\partial \mathbb{W}}{\partial \varphi} - 3 \mathbb{W} \frac{\partial^{2} \mathbb{U}}{\partial \varphi^{2}} - 3 \mathbb{W}^{2} \frac{\partial \mathbb{V}}{\partial \varphi} \\ &\quad + 6zx \frac{\partial \mathbb{U}}{\partial \varphi} + 3 \mathbb{U}^{2} \frac{\partial \mathbb{W}}{\partial \varphi}, \quad \mathfrak{F} > 0, 0 < \gamma \leq 1. \end{split}$$

The modified Korteweg-de Vries equation in its standard type is simplified by the modified couple Korteweg-de Vries equation (2), with $\mathbb{V} = \mathbb{W} = 0$. Korteweg-de Vries models are a source of nonevolution equations with a wide range of implementations in science and engineering. The Korteweg-de Vries models, for

instance, generate ion-acoustic result in fluid mechanics [21, 22]. Long waves characterise geophysical fluid dynamics in shallow and deep oceans [23, 24]. Various studies have suggested numerous systems to overcome the fractional-order Korteweg–de Vries equation employing various methodologies, such as the differential transform method [25], Adomian decomposition method [26], natural decomposition method [27], homotopy analysis method [28], Elzaki projected differential transform method [29], variational iteration method [30], new iterative method [31], modified tanh technique [32], and Lie symmetry analysis [33]. Analogously, same solutions for (2) have been suggested by Inc and Cavlak [34], Fan [35], Lin et al. [36], Inc et al. [37], and Ghoreishi et al. [18].

Daftardar-Gejji and Jafari [38] proposed an innovative iterative method of solving functional equations with approximation solutions. The new iterative approach is constructed on the justification of disappearing the nonlinear functions is identified as the iterative transformation technique [39]. This procedure is quick and accurate, and it avoids the utilization of complicated integrals, unconditioned matrix, and infinite series forms. This technique does not require any expressive parameters for the model. Numerous researchers have analyzed new iterative transformation methods to solve partial differential equations, such as the Fornberg–Whitham equation [40], KdV equation [31], and Klein–Gordon equation [41].

The Adomian decomposition method was firstly introduced by Adomian in 1980 and implemented by several investigators. In recent decades, numerous researchers have investigated the solutions of integral and differential equations by different techniques with the mixed Laplace transform. The Adomian decomposition method was modified with many integral transformations, such as Laplace, p-Laplace, Elzaki, Aboodh, and Mohand. Modification of Laplace Adomian decomposition method for solving nonlinear Volterra integral and integro-differential equations based on Newton Raphson formula [42] for solving nonlinear integrodifferential and Volterra integral equations based on the Newton-Raphson method, discrete Adomian decomposition technique [43] applied for investigating the fractional-order Navier-Stokes model, Laplace-Adomian decomposition method [44] study of implicit-impulsive differential equations involving Caputo-Fabrizio fractional derivative.

2. Basic Definitions

Definition 1. The fractional-order Caputo derivative is defined by

$${}^{\text{LC}}D_{\mathfrak{V}}^{\gamma}\left\{f\left(\mathfrak{T}\right)\right\} = \frac{1}{(n-\gamma)} \int_{0}^{\mathfrak{T}} \left(\mathfrak{T}-k\right)^{n-\gamma-1} f^{n}(k) \mathrm{d}k,$$
(3)
where $n < \gamma \le n+1.$

Definition 2. The Laplace transformation connected with fractional Caputo derivative ${}^{LC}D^{\gamma}_{\mathfrak{I}}{f(\mathfrak{I})}$ is expressed by

$$\mathbb{L}\left\{ {}^{\mathrm{LC}}D_{\mathfrak{V}}^{\gamma}\left\{f\left(\mathfrak{F}\right)\right\}\right\}(s) = \frac{1}{s^{n-\gamma}}\left[s^{n}\mathbb{L}\left\{f\left(x,\mathfrak{F}\right)\right\}\right.$$

$$(s) - s^{n-1}f\left(x,0\right) - \dots - f^{n-1}\left(x,0\right)\right].$$

$$(4)$$

Definition 3. In the Caputo sense, the Atangana-Baleanu derivative is defined as

$${}^{ABC}D^{\gamma}_{\mathfrak{F}}\{f(\mathfrak{F})\} = \frac{A(\gamma)}{1-\gamma} \int_{a}^{\mathfrak{F}} f'(k)E_{\gamma} \left[-\frac{\gamma}{1-\gamma}(1-k)^{\gamma}\right] \mathrm{d}k,$$
(5)

where $A(\gamma)$ is a normalization function such that $A(0) = A(1) = 1, f \in H^1(a, b), b > a, \gamma \in [0, 1]$, and E_{γ} represents the Mittag–Leffler function.

Definition 4. The Atangana-Baleanu derivative in the Riemann-Liouville sense is defined as

$${}^{ABC}D_{\mathfrak{F}}^{\gamma}\left\{f\left(\mathfrak{F}\right)\right\} = \frac{A\left(\gamma\right)}{1-\gamma}\frac{\mathrm{d}}{\mathrm{d}\mathfrak{F}}\int_{a}^{\mathfrak{F}}f\left(k\right)E_{\gamma}\left[-\frac{\gamma}{1-\gamma}\left(1-k\right)^{\gamma}\right]\mathrm{d}k.$$
(6)

Definition 5. The Laplace transform connected with the Atangana-Baleanu operator is defined as

$${}^{AB}D_{\mathfrak{V}}^{\gamma}\left\{f\left(\mathfrak{V}\right)\right\}(s) = \frac{A\left(\gamma\right)s^{\gamma}\mathbb{L}\left\{f\left(\mathfrak{V}\right)\right\}(s) - s^{\gamma-1}f\left(0\right)}{\left(1-\gamma\right)\left(s^{\gamma} + \left(\gamma/\left(1-\gamma\right)\right)\right)}.$$
(7)

Definition 6. Consider $0 < \gamma < 1$, and *f* is a function of γ ; then, the fractional-order integral operator of γ is given as

$${}^{ABC}I_{\mathfrak{F}}^{\gamma}\left\{f\left(\mathfrak{F}\right)\right\} = \frac{1-\gamma}{A\left(\gamma\right)}f\left(\mathfrak{F}\right) + \frac{\gamma}{A\left(\gamma\right)\Gamma\left(\gamma\right)}\int_{a}^{\mathfrak{F}}f\left(k\right)\left(\mathfrak{F}-k\right)^{\gamma-1}\mathrm{d}k.$$
(8)

3. The General Implementation of the Modified Decomposition Method

Suppose the nonlinear fractional partial differential equations

$$\mathcal{D}_{\mathfrak{F}}^{\gamma} \mathbb{U}(\varphi, \mathfrak{F}) + \mathcal{L}\mathbb{U}(\varphi, \mathfrak{F}) + \mathcal{N}\mathbb{U}(\varphi, \mathfrak{F}) = \mathcal{H}(\varphi, \mathfrak{F}), \quad \mathfrak{F} > 0, 0 < \gamma \le 1,$$
(9)

with the condition

$$\mathbb{U}(\varphi, 0) = \mathscr{G}(\varphi), \tag{10}$$

where $\mathscr{D}_{\mathfrak{T}}^{\gamma} = (\partial^{\gamma} \mathbb{U}(\varphi, \mathfrak{T})/\partial \mathfrak{T}^{\gamma})$ show the fractional-order Caputo derivative operator with $0 < \gamma \le 1$, while \mathscr{L} is linear, \mathscr{N} are nonlinear functions, and $\mathscr{H}(\varphi, \mathfrak{T})$ defines the source term.

Applying the Laplace transformation to (9), we get

$$\mathbb{L}\Big[\mathscr{D}^{\gamma}_{\mathfrak{F}}\mathbb{U}(\varphi,\mathfrak{F}) + \mathscr{L}\mathbb{U}(\varphi,\mathfrak{F}) + \mathscr{N}\mathbb{U}(\varphi,\mathfrak{F})\Big] = \mathbb{L}[\mathscr{H}(\varphi,\mathfrak{F})].$$
(11)

Taking the Laplace transformation differentiation, we find

$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\mathcal{U}\left(\nu,\omega\right) = \sum_{\kappa=0}^{j-1} \left(\frac{1}{\nu}\right)^{\gamma-\kappa-1} \mathbb{U}^{\left(\kappa\right)}\left(0\right) + \mathbb{L}[\mathcal{SU}\left(\varphi,\mathfrak{F}\right) + \mathcal{NU}\left(\varphi,\mathfrak{F}\right)] + \mathbb{L}[\mathcal{H}(\varphi,\mathfrak{F})].$$
(12)

The inverse Laplace transformation of (12) gives

$$\mathbb{U}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\sum_{\kappa=0}^{j-1} \left(\frac{1}{\nu} \right)^{\gamma-\kappa-1} \mathbb{U}^{(\kappa)}(0) + \frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{H}(\varphi, \mathfrak{F})] \right] - \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{L}\mathbb{U}(\varphi, \mathfrak{F}) + \mathscr{N}\mathbb{U}(\varphi, \mathfrak{F})] \right].$$
(13)

The adomain decomposition method series form solution is defined as

$$\mathbb{U}(\varphi,\mathfrak{F}) = \sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi,\mathfrak{F}).$$
(14)

Thus, the nonlinear function $\mathcal{N}(\varphi, \mathfrak{F})$ can be calculated by the Adomian polynomials defined as

$$\mathcal{N}\mathbb{U}(\varphi,\mathfrak{F}) = \sum_{j=0}^{\infty} \tilde{A}_j(\mathbb{U}_0,\mathbb{U}_1,\ldots), \quad j = 0, 1, \ldots,$$
(15)

$$\widetilde{A}_{j}\left(\mathbb{U}_{0},\mathbb{U}_{1},\ldots\right) = \frac{1}{j!} \left[\frac{\mathrm{d}^{j}}{\mathrm{d}\lambda^{j}} \,\mathcal{N}\left(\sum_{J=0}^{\infty} \lambda^{J} \mathbb{U}_{J}\right) \right]_{\lambda=0}, \quad j > 0.$$
(16)

Putting (14) and (15) into (13), we have

$$\sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi, \mathfrak{F}) = \mathscr{G}(\varphi) + \widetilde{\mathscr{G}}(\varphi) - \mathbb{L}^{-1} \bigg[\frac{\left(v^{\gamma}(1-\gamma)+\gamma\right)}{v^{\gamma}} \mathbb{L} \bigg]$$

$$\left[\mathscr{L}\mathbb{U}(\varphi, \mathfrak{F}) + \sum_{j=0}^{\infty} \widetilde{A}_{j} \right] \bigg].$$
(17)

Lastly, the iterative methodology for (17) is achieved as

where

(25)

$$\mathbb{U}_{0}(\varphi, \mathfrak{F}) = \mathscr{G}(\varphi) + \mathscr{G}(\varphi), \quad j = 0,$$

$$\mathbb{U}_{j+1}(\varphi, \mathfrak{F}) = -\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \left[\mathscr{L}\mathbb{U}(\varphi, \mathfrak{F}) + \sum_{j=0}^{\infty} \widetilde{A}_{j} \right] \right], \quad j \ge 1.$$
(18)

4. The General Discussion of the New Iterative **Transformation Method**

Let us assume the following general fractional partial differential equation

$$\mathcal{D}_{\mathfrak{F}}^{\gamma} \mathbb{U}(\varphi, \mathfrak{F}) + \mathcal{L}\mathbb{U}(\varphi, \mathfrak{F}) + \mathcal{N}\mathbb{U}(\varphi, \mathfrak{F}) = \mathcal{H}(\varphi, \mathfrak{F}),$$

$$\mathfrak{F} > 0, j - 1 < \gamma \le j, j \in \mathbb{N},$$
(19)

with the condition

$$\mathbb{U}^{(\kappa)}(\varphi,0) = \mathscr{G}_{\kappa}(\varphi), \quad \kappa = 0, 1, 2, \dots, j-1,$$
(20)

where \mathscr{L} and \mathscr{N} are linear and nonlinear terms and $\mathscr{H}(\varphi, \mathfrak{F})$ shows the source term.

Using the Laplace transformation to (19), we get

$$\mathbb{L}\left[\mathscr{D}^{\gamma}_{\mathfrak{F}}\mathbb{U}\left(\varphi,\mathfrak{F}\right)+\mathscr{L}\mathbb{U}\left(\varphi,\mathfrak{F}\right)+\mathscr{N}\mathbb{U}\left(\varphi,\mathfrak{F}\right)\right]=\mathbb{L}[\mathscr{H}(\varphi,\mathfrak{F})].$$
(21)

Taking the Laplace transformation differentiation property, we get

$$\frac{v^{\gamma}}{\left(v^{\gamma}\left(1-\gamma\right)+\gamma\right)}\mathcal{U}\left(v,\omega\right) = \sum_{\kappa=0}^{j-1} \left(\frac{1}{v}\right)^{\gamma-\kappa-1} \mathbb{U}^{\left(\kappa\right)}\left(0\right) + \mathbb{L}\left[\mathscr{D}\mathbb{U}\left(\varphi,\mathfrak{F}\right)+\mathscr{N}\mathbb{U}\left(\varphi,\mathfrak{F}\right)\right] + \mathbb{L}\left[\mathscr{H}(\varphi,\mathfrak{F})\right].$$
(22)

The inverse Laplace transformation of (22) gives

$$\mathbb{U}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1} \left[\sum_{\kappa=0}^{j-1} \left(\frac{1}{\nu} \right)^{\gamma-\kappa-1} \mathbb{U}^{(\kappa)}(0) + \frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{H}(\varphi,\mathfrak{F})] \right]$$

$$- \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{L}\mathbb{U}(\varphi,\mathfrak{F}) + \mathscr{N}\mathbb{U}(\varphi,\mathfrak{F})] \right].$$
(23)
ection, we achieve
$$\mathscr{L} \left(\sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi,\mathfrak{F}) \right) = \sum_{j=0}^{\infty} \mathscr{L} \left[\mathbb{U}_{j}(\varphi,\mathfrak{F}) \right],$$
(25)

From the iterative connection, we achieve

$$\mathbb{U}(\varphi, \mathfrak{F}) = \sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi, \mathfrak{F}).$$
(24)

and \mathcal{N} defines the nonlinear term as in [38].

Also, the linear operator is \mathcal{L} ; therefore,

$$\mathcal{N}\left(\sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi, \mathfrak{F})\right) = \mathcal{N}\left(\mathbb{U}_{0}(\varphi, \mathfrak{F})\right) + \sum_{j=0}^{\infty} \left[\mathcal{N}\left(\sum_{k=0}^{\infty} \mathbb{U}_{\kappa}(\varphi, \mathfrak{F})\right) - \mathcal{N}\left(\sum_{k=1}^{\infty} \mathbb{U}_{\kappa}(\varphi, \mathfrak{F})\right)\right]$$

$$= \mathcal{N}\left(\mathbb{U}_{0}\right) + \sum_{k=1}^{\infty} D_{j},$$
(26)

where $D_j = \mathcal{N}(\sum_{\kappa=0}^{j} \mathbb{U}_{\kappa}) - \mathcal{N}(\sum_{\kappa=0}^{j-1} \mathbb{U}_{\kappa}).$

By putting (24), (25), and (??) into (23), we obtain

$$\sum_{j=0}^{\infty} \mathbb{U}_{j}(\varphi, \mathfrak{F})\mathbb{L}^{-1}\left[\sum_{\kappa=0}^{j-1} \left(\frac{1}{\nu}\right)^{\gamma-\kappa-1} \mathbb{U}^{(\kappa)}(0) + \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[\mathscr{H}(\varphi, \mathfrak{F})\right]\right] -\mathbb{L}^{-1}\left\{\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[\mathscr{H}\left(\sum_{\kappa=0}^{\infty} \mathbb{U}_{\kappa}(\varphi, \mathfrak{F})\right) + \mathcal{N}(\mathbb{U}_{0}) + \sum_{\kappa=1}^{j} D_{j}\right]\right\}.$$

$$(27)$$

As a result, we determine the next iteration

$$\begin{split} \mathbb{U}_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Biggl[\sum_{\kappa=0}^{j-1} \left(\frac{1}{\nu} \right)^{\gamma-\kappa-1} \mathbb{U}^{(\kappa)}(0) + \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{H}(\varphi,\mathfrak{F})] \Biggr], \\ \mathbb{U}_{1}(\varphi,\mathfrak{F}) &= -\mathbb{L}^{-1} \Biggl\{ \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{L}(\mathbb{U}_{0}(\varphi,\mathfrak{F})) + \mathscr{N}(\mathbb{U}_{0}(\varphi,\mathfrak{F}))] \Biggr\}, \\ &\vdots \\ \mathbb{U}_{j+1}(\varphi,\mathfrak{F}) &= -\mathbb{L}^{-1} \Biggl\{ \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}[\mathscr{L}(\mathbb{U}_{j}(\varphi,\mathfrak{F})) + D_{j}] \Biggr\}, \quad m \ge 1. \end{split}$$
(28)

Finally, (19) and (20) yield the *j*-term result in the series form, defined as

$$\mathbb{U}(\varphi, \mathfrak{F}) \approx \mathbb{U}_{0}(\varphi, \mathfrak{F}) + \mathbb{U}_{1}(\varphi, \mathfrak{F}) + \mathbb{U}_{2}(\varphi, \mathfrak{F}) \\
+ \dots + \mathbb{U}_{j}(\varphi, \mathfrak{F}), \quad j \in \mathbb{N}.$$
(29)

5. Uniqueness and Existence Solutions for the Modified Decomposition Method

Theorem 1 (uniqueness theorem). The unique result of equation (9) provide space whenever $0 < \varepsilon < 1$, where $\varepsilon = (\breve{L}_1 + \breve{L}_2 + \breve{L}_3)((1 - \gamma) + (\gamma \mathfrak{T}^{\gamma}/\Gamma(\gamma + 1))).$

Proof. Assume that $J = (\mathscr{C}[I], \|.\|)$ represents all continuous mappings on the Banach space, defined on $I = [0, \mathbb{T}]$

having the norm $\|.\|$. For this, we introduce a mapping $W: M \mapsto M$, and we have

$$\mathbb{U}_{n+1}(\varphi,\mathfrak{F}) = \mathbb{U}(\varphi,\mathfrak{F}) + \mathbb{L}^{-1} \bigg[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \big[\mathscr{L} \big[\mathbb{U}_{n}(\varphi,\mathfrak{F}) \big] \\ + \mathscr{R} \big[\mathbb{U}_{n}(\varphi,\mathfrak{F}) \big] + \mathscr{N} \big[\mathbb{U}_{n}(\varphi,\mathfrak{F}) \big] \big] \big], \quad n \ge 0,$$
(30)

when $\mathscr{L}[\mathbb{U}(\varphi, \mathfrak{F})] \equiv (\partial^{3}\mathbb{U}(\varphi, \mathfrak{F})/\partial \varphi^{2})$ and $\mathscr{R}[\mathbb{U}(\varphi, \mathfrak{F})] \equiv (\partial\mathbb{U}(\varphi, \mathfrak{F})/\partial \varphi)$. Suppose that $\mathscr{L}[\mathbb{U}(\varphi, \mathfrak{F})]$ and $\mathscr{M}[\mathbb{U}(\varphi, \mathfrak{F})]$ are also Lipschitzian with $|\mathscr{R}z - \mathscr{R}\check{\mathbb{U}}| < \check{L}_{1}|\mathbb{U} - \check{\mathbb{U}}|$ and $|\mathscr{L}\mathbb{U} - \mathscr{L}\check{\mathbb{U}}| < \check{L}_{2}|\mathbb{U} - \check{\mathbb{U}}|$ where \check{L}_{1} and \check{L}_{2} are Lipschitz constants, respectively, and $\mathbb{U}, \check{\mathbb{U}}$ are various values of the mapping.

$$\|W\mathbb{U} - W\check{\mathbb{U}}\| = \max_{\mathfrak{F}\in I} \left| \begin{array}{c} \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\mathscr{L}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)] + \mathscr{R}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)] + \mathscr{N}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)]\right] \right] \right| \\ -\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\mathscr{L}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)] + \mathscr{R}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)] + \mathscr{N}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)]\right] \right] \right| \\ \leq \max_{\mathfrak{F}\in I} \left| \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\mathscr{L}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)] - \mathscr{L}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)]\right] \right] \right| \\ +\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\mathscr{R}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)] - \mathscr{R}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)]\right] \right] \right| \\ +\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\mathscr{N}[\mathbb{U}\left(\varphi,\mathfrak{F}\right)] - \mathscr{N}[\check{\mathbb{U}}\left(\varphi,\mathfrak{F}\right)]\right] \right] \right|$$

$$\leq \max_{\mathfrak{F}\in I} \left[\begin{split} \breve{L}_{1}\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\right| \right] \\ +\breve{L}_{2}\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\right| \right] \\ +\breve{L}_{3}\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\right| \right] \\ \leq \max_{\mathfrak{F}\in I} \left(\breve{L}_{1}+\breve{L}_{2}+\breve{L}_{3}\right)\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\right| \right] \\ \leq \left(\breve{L}_{1}+\breve{L}_{2}+\breve{L}_{3}\right)\mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\right| \right] \\ = \left(\breve{L}_{1}+\breve{L}_{2}+\breve{L}_{3}\right) \left(\left(1-\gamma\right)+\frac{\gamma\mathfrak{F}^{\gamma}}{\Gamma\left(\gamma+1\right)}\right) \|\mathbb{U}\left(\varphi,\mathfrak{F}\right)-\breve{U}\left(\varphi,\mathfrak{F}\right)\|. \end{split}$$

The mapping is a contraction under the assumption $0 < \varepsilon < 1$. As a result of the Banach contraction fixed point theorem, there is a unique solution to (9). As a result, the proof is complete.

Theorem 2 (convergence analysis). The solution general form of (9) will be convergent.

Proof. Suppose \widehat{S}_n is the *n*th partial sum; that is, $\widehat{W}_n = \sum_{j=0}^n \mathbb{U}_j(\varphi, \mathfrak{F})$. Firstly, we define that $\{\widehat{W}_n\}$ is a Banach space Cauchy sequence in *M*. Using into consideration of Adomian polynomials, we achieve

$$\overline{R}(\widehat{W}_n) = \breve{H}_n + \sum_{p=0}^{n-1} \breve{H}_p,$$

$$\overline{N}(\widehat{W}_n) = \breve{H}_n + \sum_{c=0}^{n-1} \breve{H}_c.$$
(32)

Now,

$$\begin{split} \left\|\widehat{W}_{n} - \widehat{W}_{q}\right\| &= \max_{\mathfrak{F} \in I} \left|\widehat{W}_{n} - \widehat{W}_{q}\right| \\ &= \max_{\mathfrak{F} \in I} \left|\sum_{j=q+1}^{n} \widetilde{U}(\varphi, \mathfrak{F})\right|, (j = 1, 2, 3, ...) \\ &\leq \max_{\mathfrak{F} \in I} \left| \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma} (1 - \gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \left[\sum_{j=q+1}^{n} \mathscr{L} \left[\mathbb{U}_{n-1} (\varphi, \mathfrak{F}) \right] \right] \right| \\ &+ \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma} (1 - \gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \left[\sum_{j=q+1}^{n} \mathscr{R} \left[\mathbb{U}_{n-1} (\varphi, \mathfrak{F}) \right] \right] \right] \\ &+ \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma} (1 - \gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \left[\sum_{j=q+1}^{n} \breve{H}_{n-1} (\varphi, \mathfrak{F}) \right] \right] \end{split}$$

$$\begin{split} &= \max_{\mathfrak{F}\in I} \left| \begin{array}{l} \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\sum_{j=q}^{n-1} \mathscr{L} \big[\mathbb{U}_{n}\left(\varphi,\mathfrak{F}\right) \big] \bigg] \right| \\ &+ \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\sum_{j=q}^{n-1} \mathscr{R} \big[\mathbb{U}_{n}\left(\varphi,\mathfrak{F}\right) \big] \bigg] \right| \\ &+ \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\sum_{j=q}^{n-1} \mathscr{L} \left(\widehat{W}_{n-1}\right) - \mathscr{L} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &\leq \max_{\mathfrak{F}\in I} \left| + \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\sum_{j=q}^{n-1} \mathscr{R} \left(\widehat{W}_{n-1}\right) - \mathscr{R} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &+ \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\mathscr{L} \left(\widehat{W}_{n-1}\right) - \mathscr{N} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &\leq \max_{\mathfrak{F}\in I} \left| + \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\mathscr{L} \left(\widehat{W}_{n-1}\right) - \mathscr{L} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &\leq \max_{\mathfrak{F}\in I} \left| + \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\mathscr{R} \left(\widehat{W}_{n-1}\right) - \mathscr{N} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &+ \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\left(\widehat{W} \left(\widehat{W}_{n-1}\right) - \mathscr{N} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \\ &= \operatorname{Max}_{\mathfrak{F}\in I} \left| \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\left(\widehat{W} \left(\widehat{W}_{n-1}\right) - \mathcal{N} \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \right| \\ &= \operatorname{Max}_{\mathfrak{F}\in I} \left| \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \bigg[\left(\widehat{W}_{n-1}\right) - \left(\widehat{W}_{q-1}\right) \bigg] \bigg] \right| \end{aligned}$$

$$+ \breve{L}_{3} \max_{\mathfrak{F}\in I} \left\| \left[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \left[\left(\widehat{W}_{n-1} \right) - \left(\widehat{W}_{q-1} \right) \right] \right\|$$
$$= \left(\breve{L}_{1} + \breve{L}_{2} + \breve{L}_{3} \right) \left((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) \left\| \widehat{W}_{n-1} - \widehat{W}_{q-1} \right\|.$$
(33)

Consider n = q + 1; then,

$$\begin{split} \left\| \widehat{W}_{q+1} - \widehat{W}_{q} \right\| &\leq \varepsilon \left\| \widehat{W}_{q} - \widehat{W}_{q-1} \right\| \\ &\leq \varepsilon^{2} \left\| \widehat{W}_{q-1} - \widehat{W}_{q-2} \right\| \leq \cdots \leq \varepsilon^{q} \left\| \widehat{W}_{1} - \widehat{W}_{0} \right\|, \end{split}$$
(34)

where $((\breve{L}_1 + \breve{L}_2 + \breve{L}_3)\mathfrak{F}^{(\gamma-1)}/\gamma!)$. Similarly, we have the triangular inequality

$$\begin{split} \left\|\widehat{W}_{n}-\widehat{W}_{q}\right\| &\leq \left\|\widehat{W}_{q+1}-\widehat{W}_{q}\right\| + \left\|\widehat{W}_{q+2}-\widehat{W}_{q+1}\right\| \\ &\quad + \dots + \left\|\widehat{W}_{n}-\widehat{W}_{n-1}\right\| \\ &\leq \left[\varepsilon^{q}+\varepsilon^{q+1}+\dots+\varepsilon^{n-1}\right] \left\|\widehat{W}_{1}-\widehat{W}_{0}\right\| \\ &\leq \varepsilon^{q} \left(\frac{1-\varepsilon^{n-q}}{\varepsilon}\right) \left\|\mathbb{U}_{1}\right\|, \end{split}$$
(35)

and since $0 < \varepsilon < 1$, we get $(1 - \varepsilon^{n-q}) < 1$; then,

$$\left\|\widehat{W}_{n} - \widehat{W}_{q}\right\| \leq \frac{\varepsilon^{q}}{1 - \varepsilon} \max_{\Im \in I} \left\|\mathbb{U}_{1}\right\|.$$
(36)

However, $|\mathbb{U}_1| < \infty$ (since $\mathbb{U}(\varphi, \mathfrak{F})$ is bounded). Thus, as $q \mapsto \infty$, $\|\widehat{W}_n - \widehat{W}_q\| \mapsto 0$. Hence, $\{\widehat{W}_1\}$ is a Cauchy sequence in *K*. As a solution, the series $\sum_{n=0}^{\infty} \mathbb{U}_n$ converges, and this completes the proof.

Theorem 3 (error estimate). *The maximum absolute truncation error of series solution (9) to (??) is computed as*

$$\max_{\mathfrak{F}\in I} \left| \mathbb{U}(\varphi,\mathfrak{F})\sum_{n=1}^{q} \mathbb{U}_{n}(\varphi,\mathfrak{F}) \right| \leq \frac{\varepsilon^{q}}{1-\varepsilon} \max_{\mathfrak{F}\in I} \left\| \mathbb{U}_{1} \right\|.$$
(37)

6. Numerical Results

This section describes several test examples by applying two novel techniques, modified decomposition technique and new iterative transformation technique, via the Atangana–Baleanu derivative operator. Also, the stability and convergence of the technique are discussed.

Example 1 (see [31]). Consider the fractional-order nonlinear system of Korteweg–de Vries equation (1) with $\vartheta = \rho = 1$, with the initial conditions

$$U(\varphi, 0) = \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \right),$$

$$W(\varphi, 0) = \sqrt{\frac{\rho}{2}} \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \right).$$
(38)

Case I: first, we apply the modified decomposition technique for Example 1.

Applying the Laplace transform to (1), we get

$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\left\{\mathcal{U}\left(\varphi,\nu\right)-\sum_{\kappa=0}^{j-1}\left(\frac{1}{\nu}\right)^{\gamma-\kappa-1}\mathbb{U}^{(\kappa)}\left(0\right)\right\}$$
$$=\mathbb{L}\left[-\rho\frac{\partial^{3}\mathbb{U}}{\partial\varphi^{3}}-6\rho\mathbb{U}\frac{\partial\mathbb{U}}{\partial\varphi}+6\mathbb{V}\frac{\partial\mathbb{V}}{\partial\varphi}\right],$$
$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\left\{\mathcal{V}\left(\varphi,\nu\right)-\sum_{\kappa=0}^{j-1}\left(\frac{1}{\nu}\right)^{\gamma-\kappa-1}\mathbb{V}^{(\kappa)}\left(0\right)\right\}$$
$$=\mathbb{L}\left[-\rho\frac{\partial^{3}\mathbb{V}}{\partial\varphi^{3}}-3\rho\mathbb{U}\frac{\partial\mathbb{V}}{\partial\varphi}\right].$$
(39)

In view of (38) and analytical method procedure as follows:

$$\mathscr{U}(\varphi, \nu) = \frac{1}{\nu} \mathbb{U}^{(0)}(\varphi, 0) + \frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}$$
$$\cdot \left[-\rho \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - 6\rho \mathbb{U} \frac{\partial \mathbb{U}}{\partial \varphi} + 6\mathbb{V} \frac{\partial \mathbb{V}}{\partial \varphi}\right], \tag{40}$$
$$\mathscr{V}(\varphi, \nu) = \frac{1}{\nu} \mathbb{V}^{(0)}(\varphi, 0) + \frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}$$
$$\cdot \left[-\rho \frac{\partial^{3} \mathbb{V}}{\partial \varphi^{3}} - 3\rho \mathbb{U} \frac{\partial \mathbb{V}}{\partial \varphi}\right].$$

Using the inverse Laplace transformation, we get

$$\mathbb{U}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{U}(\varphi, 0) \Big] + \mathbb{L}^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \\
\cdot \Big[-\rho \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - 6\rho \mathbb{U} \frac{\partial \mathbb{U}}{\partial \varphi} + 6\mathbb{V} \frac{\partial \mathbb{V}}{\partial \varphi} \Big] \Big], \\
\mathbb{V}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{V}(\varphi, 0) \Big] + \mathbb{L}^{-1} \\
\cdot \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \Big[-\rho \frac{\partial^{3} \mathbb{V}}{\partial \varphi^{3}} - 3\rho \mathbb{U} \frac{\partial \mathbb{V}}{\partial \varphi} \Big] \Big]. \tag{41}$$

By morality of the modified decomposition technique, we get

$$\begin{split} \mathbb{U}_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{\omega}{\nu} \mathbb{U}(\varphi,0) \right] \\ &= \mathbb{L}^{-1} \left[\frac{1}{\nu} \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \right) \right] \\ &= \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \right), \\ \mathbb{V}_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{\omega}{\nu} \mathbb{V}(\varphi,0) \right] \\ &= \sqrt{\frac{\rho}{2}} \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \right), \\ \sum_{j=0}^{\infty} \mathbb{U}_{j+1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \right] \\ &\quad \cdot \left[-\rho \sum_{j=0}^{\infty} \left(\mathbb{U}_{\varphi\varphi\varphi} \right)_{j} - 6\rho \sum_{j=0}^{\infty} \mathscr{A}_{j} + 6 \sum_{j=0}^{\infty} \mathscr{B}_{j} \right] \right], \\ \sum_{j=0}^{\infty} \mathbb{V}_{j+1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \right] \\ &\quad \cdot \left[-\rho \sum_{j=0}^{\infty} \left(\mathbb{V}_{\varphi\varphi\varphi} \right)_{j} - 3\rho \sum_{j=0}^{\infty} \mathscr{C}_{j} \right] \end{split}$$

The Adomian polynomials' some terms are defined as follows:

$$\begin{aligned} \mathscr{A}_{0}\left(\mathbb{U}\mathbb{U}_{\varphi}\right) &= \mathbb{U}_{0}\mathbb{U}_{0\varphi}, \\ \mathscr{A}_{1}\left(\mathbb{U}\mathbb{U}_{\varphi}\right) &= \mathbb{U}_{0}\mathbb{U}_{1\varphi} + \mathbb{U}_{1}\mathbb{U}_{0\varphi}, \\ \mathscr{A}_{2}\left(\mathbb{U}\mathbb{U}_{\varphi}\right) &= \mathbb{U}_{1}\mathbb{U}_{2\varphi} + \mathbb{U}_{1}\mathbb{U}_{1\varphi} + \mathbb{U}_{2}\mathbb{U}_{0\varphi}, \\ \mathscr{B}_{0}\left(\mathbb{V}\mathbb{V}_{\varphi}\right) &= \mathbb{V}_{0}\mathbb{V}_{0\varphi}, \\ \mathscr{B}_{1}\left(\mathbb{V}\mathbb{V}_{\varphi}\right) &= \mathbb{V}_{0}\mathbb{V}_{1\varphi} + \mathbb{V}_{1}\mathbb{V}_{0\varphi}, \\ \mathscr{B}_{2}\left(\mathbb{V}\mathbb{V}_{\varphi}\right) &= \mathbb{V}_{1}\mathbb{V}_{2\varphi} + \mathbb{V}_{1}\mathbb{V}_{1\varphi} + \mathbb{V}_{2}\mathbb{V}_{0\varphi}, \\ \mathscr{C}_{0}\left(\mathbb{U}\mathbb{V}_{\varphi}\right) &= \mathbb{U}_{0}\mathbb{V}_{0\varphi}, \\ \mathscr{C}_{1}\left(\mathbb{U}\mathbb{V}_{\varphi}\right) &= \mathbb{U}_{0}\mathbb{V}_{1\varphi} + \mathbb{U}_{1}\mathbb{V}_{0\varphi}, \\ \mathscr{C}_{2}\left(\mathbb{U}\mathbb{V}_{\varphi}\right) &= \mathbb{U}_{1}\mathbb{V}_{2\varphi} + \mathbb{U}_{1}\mathbb{V}_{1\varphi} + \mathbb{U}_{2}\mathbb{V}_{0\varphi}. \end{aligned}$$

For $j = 0, 1, 2, 3, \ldots$,

2)

$$\begin{split} \mathbb{U}_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \Big[-\rho \Big(\mathbb{U}_{\varphi\varphi\varphi}\Big)_{0} - 6\rho\mathcal{A}_{0} + 6\mathcal{B}_{0} \Big] \bigg] \\ &= \mathbb{L}^{-1} \bigg[\frac{\omega^{\gamma+2}}{\nu^{\gamma+2}} \varrho^{5} \rho \, \tanh \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \sec h^{2} \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \bigg] \\ &= \varrho^{5} \rho \, \tanh \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \sec h^{2} \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \bigg((1-\gamma) + \frac{\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)} \bigg) \\ \mathbb{V}_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \Big[-\rho \big(\mathbb{V}_{\varphi\varphi\varphi}\Big)_{0} - 3\rho\mathcal{C}_{0} \Big] \bigg] \\ &= \frac{\varrho^{5} \rho^{3/2}}{\sqrt{2}} \tanh \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \sec h^{2} \bigg(\frac{\delta}{2} + \frac{\varrho\varphi}{2} \bigg) \bigg((1-\gamma) + \frac{\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)} \bigg) \\ \mathbb{U}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L} \Big[-\rho \big(\mathbb{U}_{\varphi\varphi\varphi}\Big)_{1} - 6\rho\mathcal{A}_{1} + 6\mathcal{B}_{1} \big] \bigg] \end{split}$$

$$= \mathbb{L}^{-1} \left[\frac{\omega^{2\gamma+2}}{\nu^{2\gamma+2}} \frac{\varrho^{8} \rho^{2}}{2} \left[2 \cosh^{2} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) - 3 \right] \sec h^{4} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) \right]$$

$$= \frac{\varrho^{8} \rho^{2}}{2} \left[2 \cosh^{2} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) - 3 \right] \sec h^{4} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right)$$

$$\mathbb{V}_{2}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1 - \gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \left[-\rho \left(\mathbb{V}_{\varphi \varphi \varphi} \right)_{1} - 3\rho \mathscr{C}_{1} \right] \right]$$

$$= \frac{\varrho^{5} \rho^{5/2}}{2\sqrt{2}} \left[2 \cosh^{2} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) - 3 \right] \sec h^{4} \left(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \right) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right)$$

$$\vdots .$$

$$(44)$$

The modified decomposition technique result for Example 1 is shown as

$$\mathbb{U}(\varphi,\mathfrak{F}) = \mathbb{U}_{0}(\varphi,\mathfrak{F}) + \mathbb{U}_{1}(\varphi,\mathfrak{F}) + \mathbb{U}_{2}(\varphi,\mathfrak{F}) + \mathbb{U}_{3}(\varphi,\mathfrak{F}) + \cdots$$

$$= \varrho^{2}\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right) + \varrho^{5}\rho \tanh\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma) + \frac{\gamma\mathfrak{F}}{\Gamma(\gamma+1)}\right)$$

$$+ \frac{\varrho^{8}\rho^{2}}{2}\left[2\cosh^{2}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right) - 3\right]\operatorname{sec} h^{4}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma)^{2} + \frac{\gamma^{2}\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)}\right) + \cdots$$

$$(45)$$

Similarly, we get

$$\mathbb{V}(\varphi,\mathfrak{F}) = \sqrt{\frac{\rho}{2}}\varrho^{2}\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right) + \frac{\varrho^{5}\rho^{3/2}}{\sqrt{2}}\operatorname{tanh}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma) + \frac{\gamma\mathfrak{F}}{\Gamma(\gamma+1)}\right) + \frac{\varrho^{5}\rho^{5/2}}{2\sqrt{2}}\left[2\operatorname{cosh}^{2}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right) - 3\right]\operatorname{sec} h^{4}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma)^{2} + \frac{\gamma^{2}\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)}\right) + \cdots \right)$$

$$(46)$$

By putting $\gamma = 1$, we achieve the exact result of the system of Korteweg–de Vries equation (1):

$$\mathbb{U}(\varphi,\mathfrak{F}) = \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho\varphi}{2} - \frac{\rho\varrho^{3}\mathfrak{F}}{2} \right),$$

$$\mathbb{V}(\varphi,\mathfrak{F}) = \sqrt{\frac{\rho}{2}} \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho\varphi}{2} - \frac{\rho\varrho^{3}\mathfrak{F}}{2} \right).$$
(47)

Case II: now, we apply the new iterative transformation technique on Example 1.

Using the suggested analytical method, we have

$$\begin{split} U_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{1}{\nu} U(\varphi,0) \Big] = \mathbb{L}^{-1} \Big[\frac{1}{\nu} q^{2} \sec h^{2} \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \Big] \\ &= \varrho^{2} \sec h^{2} \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \\ V_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{0}}{\partial \varphi^{2}} - 6\rho \mathbb{L}_{0} \frac{\partial U_{0}}{\partial \varphi} + 6V_{0} \frac{\partial V_{0}}{\partial \varphi} \Big] \Big] \\ &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} q^{5} \rho \tanh \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \sec h^{2} \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \Big] \\ &= \varrho^{5} \rho \tanh \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \sec h^{2} \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \Big((1-\gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ V_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{0}}{\partial \varphi^{2}} - 3\rho \mathbb{U}_{0} \frac{\partial V_{0}}{\partial \varphi} \Big] \Big] \\ &= \frac{\varrho^{5} \rho^{3/2}}{\sqrt{2}} \tanh \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \sec h^{2} \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) \Big((1-\gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ V_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{1}}{\partial \varphi^{2}} - 6\rho \mathbb{U}_{1} \frac{\partial U_{1}}{\partial \varphi} + 6V_{1} \frac{\partial V_{1}}{\partial \varphi} \Big] \Big] \\ &= \frac{\varrho^{5} \rho^{3/2}}{\sqrt{2}} \tanh \Big(\frac{\delta}{2} + \frac{\varrho \varphi}{2} \Big) = \delta h^{3} \Big(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \Big) \Big((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{S}^{2} r}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma) r \mathfrak{S}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ U_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{1}}{\partial \varphi^{2}} - 3\rho \mathbb{U}_{1} \frac{\partial V_{1}}{\partial \varphi} \Big] \Big] \\ &= \frac{\varrho^{5} \rho^{3/2}}{2} \Big[2 \cosh^{2} \Big(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \Big) - 3 \Big] \sec h^{4} \Big(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \Big) \Big((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{S}^{2} r}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma) r \mathfrak{S}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ &\vdots \\ U_{n}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{1}}{\partial \varphi^{2}} - 3\rho \mathbb{U}_{1} \frac{\partial V_{1}}{\partial \varphi} \Big] \Big] \\ &= \frac{\varrho^{5} \rho^{5/2}}{2\sqrt{2}} \Big[2 \cosh^{2} \Big(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \Big) - 3 \Big] \sec h^{4} \Big(\frac{\rho}{2} + \frac{\varrho \varphi}{2} \Big) \Big((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{S}^{2} r}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma) r \mathfrak{S}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ &\vdots \\ U_{n}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \Big[-\rho \frac{\partial^{3} U_{1}}{\partial \varphi^{2}} - 3\rho \mathbb{U}_{1} - \frac{\partial^{3} U_{1}}{\partial \varphi} + 6V_{1} - \frac{\partial^{3} V_{1}}{\partial \varphi} \Big] \Big] \\ \end{bmatrix}$$

The series of solutions for Example 1 is expressed as

Consequently, we have

$$\mathbb{U}(\varphi, \mathfrak{F}) = \mathbb{U}_{0}(\varphi, \mathfrak{F}) + \mathbb{U}_{1}(\varphi, \mathfrak{F}) + \mathbb{U}_{2}(\varphi, \mathfrak{F}) \\
+ \mathbb{U}_{3}(\varphi, \mathfrak{F}) + \cdots \mathbb{U}_{j}(\varphi, \mathfrak{F}), \\
\mathbb{V}(\varphi, \mathfrak{F}) = \mathbb{V}_{0}(\varphi, \mathfrak{F}) + \mathbb{V}_{1}(\varphi, \mathfrak{F}) + \mathbb{V}_{2}(\varphi, \mathfrak{F}) \\
+ \mathbb{V}_{3}(\varphi, \mathfrak{F}) + \cdots \mathbb{V}_{j}(\varphi, \mathfrak{F}).$$
(49)

$$\mathbb{U}(\varphi,\mathfrak{F}) = \varrho^{2}\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right) + \varrho^{5}\rho \tanh\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma) + \frac{\gamma\mathfrak{F}}{\Gamma(\gamma+1)}\right) \\ + \frac{\varrho^{8}\rho^{2}}{2}\left[2\cosh^{2}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right) - 3\right]\operatorname{sec} h^{4}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma)^{2} + \frac{\gamma^{2}\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)}\right) + \cdots,$$

$$\mathbb{V}(\varphi,\mathfrak{F}) = \sqrt{\frac{\rho}{2}}\varrho^{2}\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right) + \frac{\varrho^{5}\rho^{3/2}}{\sqrt{2}}\tanh\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\operatorname{sec} h^{2}\left(\frac{\delta}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma) + \frac{\gamma\mathfrak{F}}{\Gamma(\gamma+1)}\right) \\ + \frac{\varrho^{5}\rho^{5/2}}{2\sqrt{2}}\left[2\cosh^{2}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right) - 3\right]\operatorname{sec} h^{4}\left(\frac{\rho}{2} + \frac{\varrho\varphi}{2}\right)\left((1-\gamma)^{2} + \frac{\gamma^{2}\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{F}^{\gamma}}{\Gamma(\gamma+1)}\right) + \cdots.$$

$$(50)$$

By putting $\gamma = 1$, we get the exact result of the system of Korteweg–de Vries equation (1):

$$\mathbb{U}(\varphi, \mathfrak{F}) = \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho \varphi}{2} - \frac{\rho \varrho^{3} \mathfrak{F}}{2} \right),$$

$$\mathbb{V}(\varphi, \mathfrak{F}) = \sqrt{\frac{\rho}{2}} \varrho^{2} \sec h^{2} \left(\frac{\delta}{2} + \frac{\varrho \varphi}{2} - \frac{\rho \varrho^{3} \mathfrak{F}}{2} \right).$$
(51)

In Figures 1 and 2, the actual and analytical solutions of $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$ are proved at $\delta = 2, \rho = 0.5$, and $\varrho = 1$. In Figures 3 and 4, the surface and two-dimensional figure for $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$ for numerous fractional orders are described which demonstrate that the modified decomposition technique and new iterative transformation technique obtained series form solutions are in close contact with the analytical and the exact results. This comparison shows a strong connection among the modified decomposition method and actual solutions. Consequently, the modified decomposition technique and new iterative transformation technique are accurate innovative techniques which need less calculation time and are very simple and more flexible than the homotopy analysis technique and homotopy perturbation technique.

Example 2 (see [31]). Consider the fractional-order nonlinear system of Korteweg-de Vries equation given as

$$\frac{\partial^{\gamma} \mathbb{U}}{\partial \mathfrak{F}^{\gamma}} = -\frac{\partial \mathbb{V}}{\partial \varphi} - \frac{1}{2} \frac{\partial \mathbb{U}^{2}}{\partial \varphi}, \qquad (52)$$

$$\frac{\partial^{\gamma} \mathbb{V}}{\partial \mathfrak{F}^{\gamma}} = -\frac{\partial \mathbb{U}}{\partial \varphi} - \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - \frac{\partial zy}{\partial \varphi}, \qquad \mathfrak{F} > 0, 0 < \gamma \le 1,$$

with the conditions

$$\mathbb{U}(\varphi, 0) = \rho \left[\tanh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) + 1 \right],$$

$$\mathbb{V}(\varphi, 0) = \frac{\rho^2}{2} \sec h^2 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) - 1.$$
(53)

Case I: first, we apply the modified decomposition technique for Example 2.

Applying the Laplace transform to (52), we find

$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\mathcal{U}\left(\varphi,\nu\right)-\sum_{\kappa=0}^{j-1}\left(\frac{1}{\nu}\right)^{\gamma-\kappa-1}\mathbb{U}^{(\kappa)}\left(0\right)=\mathbb{E}\left[-\frac{\partial\mathbb{V}}{\partial\varphi}-\frac{1}{2}\frac{\partial\mathbb{U}^{2}}{\partial\varphi}\right],$$

$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\mathcal{V}\left(\varphi,\nu\right)-\sum_{\kappa=0}^{j-1}\left(\frac{1}{\nu}\right)^{\gamma-\kappa-1}\mathbb{V}^{(\kappa)}\left(0\right)=\mathbb{E}\left[-\frac{\partial\mathbb{U}}{\partial\varphi}-\frac{\partial^{3}\mathbb{U}}{\partial\varphi^{3}}-\frac{\partial\mathrm{zy}}{\partial\varphi}\right].$$
(54)

In view of (29) and straightforward approximate achieve

$$\mathcal{U}(\varphi, \nu) = \frac{1}{\nu} \mathbb{U}^{(0)}(\varphi, 0) + \frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L}\left[-\frac{\partial \mathbb{V}}{\partial \varphi} - \frac{1}{2} \frac{\partial \mathbb{U}^{2}}{\partial \varphi}\right],$$

$$\mathcal{V}(\varphi, \nu) = \frac{1}{\nu} \mathbb{V}^{(0)}(\varphi, 0) + \frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L}$$

$$\cdot \left[-\frac{\partial \mathbb{U}}{\partial \varphi} - \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - \frac{\partial zy}{\partial \varphi}\right].$$
(55)

Using the inverse Laplace transformation, we get

$$\mathbb{U}(\varphi, \mathfrak{T}) = \mathbb{L}^{-1} \left[\frac{1}{\nu} \mathbb{U}(\varphi, 0) \right] + \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \right]$$
$$\cdot \left[-\frac{\partial \mathbb{V}}{\partial \varphi} - \frac{1}{2} \frac{\partial \mathbb{U}^{2}}{\partial \varphi} \right],$$
$$\mathbb{V}(\varphi, \mathfrak{T}) = \mathbb{L}^{-1} \left[\frac{1}{\nu} \mathbb{V}(\varphi, 0) \right] + \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \right]$$
$$\cdot \left[-\frac{\partial \mathbb{U}}{\partial \varphi} - \frac{\partial^{3} \mathbb{U}}{\partial \varphi^{3}} - \frac{\partial zy}{\partial \varphi} \right].$$
(56)

By the consequence of the modified decomposition technique, we get

$$\begin{split} \mathbb{U}_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{U}(\varphi,0) \Big] \\ &= \mathbb{L}^{-1} \Big[\frac{1}{\nu} \rho \Big(\tanh \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) + 1 \Big) \Big] \\ &= \rho \Big(\tanh \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) + 1 \Big), \end{split}$$
(57)
$$\\ \mathbb{V}_{0}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{V}(\varphi,0) \Big] \end{split}$$

 $\mathbb{U}_{1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[-\rho\left(\mathbb{V}_{\varphi}\right)_{0}-\frac{1}{2}\mathcal{D}_{0}\right]\right]$

 $= -\frac{\rho^2}{2} \mathbb{L}^{-1} \left[\frac{\omega^{\gamma+2}}{v^{\gamma+2}} \sec h^2 \left(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \right) \right]$

 $= -\frac{\rho^2}{2} \sec h^2 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \left((1-\gamma) + \frac{\gamma \mathfrak{T}^{\gamma}}{\Gamma(\gamma+1)}\right)$

 $\mathbb{V}_{1}(\varphi,\mathfrak{T}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}(1-\gamma)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[-\left(\mathbb{U}_{\varphi}\right)_{0}-\left(\mathbb{V}_{\varphi\varphi\varphi\phi}\right)_{0}-\left((zy)_{\varphi}\right)_{0}\right]\right]$

It follows that

$$\sum_{j=0}^{\infty} \mathbb{U}_{j+1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma} (1-\gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \right]$$
$$\cdot \left[-\rho \sum_{j=0}^{\infty} (\mathbb{V}_{\varphi})_{j} - \frac{1}{2} \sum_{j=0}^{\infty} \mathcal{D}_{j} \right],$$
$$\sum_{j=0}^{\infty} \mathbb{V}_{j+1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma} (1-\gamma) + \gamma)}{\nu^{\gamma}} \mathbb{L} \right]$$
$$\cdot \left[-\sum_{j=0}^{\infty} (\mathbb{U}_{\varphi})_{j} - \sum_{j=0}^{\infty} (\mathbb{V}_{\varphi\varphi\varphi\phi})_{j} - \sum_{j=0}^{\infty} (\mathbb{U}\mathbb{V}_{\varphi})_{j} \right],$$
$$j = 0, 1, 2, \dots.$$
(58)

The Adomian polynomials' some terms are expressed as

$$\mathcal{D}_{0}(\mathbb{U}^{2}) = \mathbb{U}_{0}^{2},$$

$$\mathcal{D}_{1}(\mathbb{U}^{2}) = 2\mathbb{U}_{0}\mathbb{U}_{1},$$

$$\mathcal{D}_{2}(\mathbb{U}^{2}) = 2\mathbb{U}_{0}\mathbb{U}_{2} + \mathbb{U}_{1}^{2}.$$
(59)

For j = 0, 1, 2, ...,

$$\begin{split} &= \frac{\rho^3}{2} \sinh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^3\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \left((1-\gamma) + \frac{\gamma\mathfrak{V}}{\Gamma(\gamma+1)}\right) \\ & \mathbb{U}_2(\varphi, \mathfrak{V}) = \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \left[-\rho(\mathbb{V}_{\varphi})_1 - \frac{1}{2}\mathcal{D}_1\right]\right] \\ &= \mathbb{L}^{-1} \left[-\frac{\rho^5}{4} \sec h^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) + \frac{3\rho^5}{4} \frac{\omega^{2\gamma+2}}{\nu^{2\gamma+2}} \sinh^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^4\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right)\right] \\ &+ \frac{\rho^7}{4} \mathbb{L}^{-1} \left[\frac{\Gamma(2\gamma+1)}{\Gamma^2(\gamma+1)} \frac{\omega^{3\gamma+2}}{\nu^{3\gamma+2}} \sinh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^5\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right)\right] \\ &= \left[-\frac{\rho^5}{4} \sec h^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) + \frac{3\rho^5}{4} \sinh^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^4\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right)\right] \left((1-\gamma) + \frac{\gamma\mathfrak{V}}{\Gamma(\gamma+1)}\right) \\ &+ \frac{2(1-\gamma)\gamma\mathfrak{V}}{\Gamma(\gamma+1)} + \frac{\rho^7}{4} \sinh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^5\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \frac{\Gamma(2\gamma+1)\mathfrak{S}^{3\gamma}}{\Gamma^2(\gamma+1)\Gamma(3\gamma+1)} \\ &\mathbb{V}_2(\varphi, \mathfrak{V}) = \mathbb{L}^{-1} \left[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \left[-(\mathbb{U}_{\varphi})_1 - (\mathbb{V}_{\varphi\varphi\varphi})_1 - ((zy)_{\varphi})_1\right]\right] \\ &= \frac{\rho^6}{4} \left[2\cosh^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) - 3\right] \sec h^4\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \left((1-\gamma)^2 + \frac{\gamma^2\mathfrak{V}^2}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{V}}{\Gamma(\gamma+1)}\right) \\ &\vdots . \end{split}$$

The modified decomposition technique result for Example 2 is represented as

$$\begin{split} \mathbb{U}(\varphi,\mathfrak{F}) &= \mathbb{U}_{0}(\varphi,\mathfrak{F}) + \mathbb{U}_{1}(\varphi,\mathfrak{F}) + \mathbb{U}_{2}(\varphi,\mathfrak{F}) + \cdots, \\ &= \rho \Big(\tanh \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) + 1 \Big) - \frac{\rho^{2}}{2} \sec h^{2} \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) \Big((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \Big) \\ &+ \Big[-\frac{\rho^{5}}{4} \sec h^{2} \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) + \frac{3\rho^{5}}{4} \sinh^{2} \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{4} \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) \Big] \Big((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \Big) \\ &+ \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{\rho^{7}}{4} \sinh \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{5} \Big(\frac{\varrho}{2} + \frac{\rho \varphi}{2} \Big) \frac{\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma^{2}(\gamma + 1)\Gamma(3\gamma + 1)} + \cdots. \end{split}$$
(61)

Consequently, we get

$$\mathbb{V}(\varphi,\mathfrak{F}) = -1 + \frac{\rho^2}{2} \sec h^2 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) + \frac{\rho^3}{2} \sinh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \sec h^3 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \left((1-\gamma) + \frac{\gamma\mathfrak{F}}{\Gamma(\gamma+1)}\right) + \frac{\rho^6}{\Gamma(\gamma+1)} \left(2\cosh^2\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) - 3\right) \sec h^4 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2}\right) \left((1-\gamma)^2 + \frac{\gamma^2\mathfrak{F}^2\gamma}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma\mathfrak{F}^\gamma}{\Gamma(\gamma+1)}\right) + \cdots$$
(62)

By putting $\gamma = 1$, we achieve the exact result of the system of Korteweg–de Vries equation:

$$\mathbb{U}(\varphi,\mathfrak{F}) = \rho \left(\tanh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2} - \frac{\rho^{2}\mathfrak{F}}{2}\right) + 1 \right),$$

$$\mathbb{V}(\varphi,\mathfrak{F}) = \frac{\rho^{2}}{2} \sec h^{2} \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2} - \frac{\rho^{2}\mathfrak{F}}{2}\right) - 1.$$
(63)

Case II: now, we implement the new iterative transformation technique on Example 2.

By using the suggested analytical technique, we get

$$\begin{split} & U_{0}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{1}{\nu} U(\varphi,0) \Big] = L^{-1} \Big[\frac{1}{\nu} \rho \Big(\tanh \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) + 1 \Big) \Big] \\ &= \rho \Big(\tanh \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) + 1 \Big) \\ & V_{0}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{1}{\nu} V(\varphi,0) \Big] \\ & U_{1}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac{\partial V_{0}}{\partial \varphi} - \frac{1}{2} \frac{\partial U_{0}^{2}}{\partial \varphi} \Big] \Big] \\ &= \frac{\rho^{2}}{2} L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac{\partial U_{0}}{\partial \varphi} - \frac{\partial^{3} U_{0}}{\partial \varphi^{2}} - \frac{\partial U_{0} V_{0}}{\partial \varphi} \Big] \Big] \\ &= \frac{\rho^{2}}{2} \sec h^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big((1-\gamma) + \frac{\gamma \mathfrak{R}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ & V_{1}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac{\partial U_{0}}{\partial \varphi} - \frac{\partial^{3} U_{0}}{\partial \varphi^{2}} - \frac{\partial U_{0} V_{0}}{\partial \varphi} \Big] \Big] \\ &= \frac{\rho^{2}}{2} \sinh \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{3} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big((1-\gamma) + \frac{\gamma \mathfrak{R}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ & U_{2}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac{\partial V_{0}}{\partial \varphi} - \frac{1}{2} \frac{\partial U_{0}^{2}}{\partial \varphi} \Big] \Big] \\ &= L^{-1} \Big[-\frac{\rho^{2}}{4} \sec h^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) + \frac{3\rho^{5}}{4} \frac{\delta^{\mu/\gamma}}{\nu^{\gamma/2}} \sinh \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{4} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big] \\ &+ \frac{\rho^{2}}{4} L^{-1} \Big[\frac{\Gamma(2\gamma+1)}{\nu^{2}} \frac{\partial^{3}\gamma}{\nu^{2}} + \frac{\delta \eta^{5}}{4} \sinh \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{4} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big] \Big((1-\gamma) + \frac{\gamma \mathfrak{R}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ &+ \frac{2(1-\gamma)^{\gamma} \mathfrak{R}^{\gamma}}{\Gamma(\gamma+1)} \frac{\partial^{3}\gamma^{2}}{\partial^{2}} - \frac{\partial^{3}}{2} \sin h^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \sec h^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \frac{\Gamma(2\gamma+1)\mathfrak{R}^{3\gamma}}{\Gamma(\gamma+1)} \Big) \\ &+ \frac{2(1-\gamma)^{\gamma} \mathfrak{R}^{\gamma}}{\mu^{2}} \Big] \Big[-\frac{\partial^{3}}{\partial \varphi} - \frac{\partial^{3}}{2} - \frac{\partial^{3}}{2} \Big] \Big] \\ &= \frac{\rho^{6}}{4} \Big[2 \cosh^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) - 3 \Big] \sec h^{4} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{R}^{3}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)^{\gamma} \mathfrak{R}^{\gamma}}{V} \Big) \\ &\vdots \\ U_{j}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac{\partial U_{j}}{\partial \varphi} - \frac{1}{2} \frac{\partial U_{j}^{2}}{\partial \varphi} \Big] \Big] \\ \\ &= \frac{\rho^{6}}{4} \Big[2 \cosh^{2} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) - 3 \Big] \sec h^{4} \Big(\frac{\vartheta}{2} + \frac{\rho \varphi}{2} \Big) \Big((1-\gamma)^{2} + \frac{\chi^{2} \mathfrak{R}^{3}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)^{\gamma} \mathfrak{R}^{\gamma}}{\Gamma(\gamma+1)} \Big) \\ &\vdots \\ U_{j}(\varphi,\mathfrak{F}) = L^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} U_{1} \Big[-\frac$$

Consequently, we have

The series of results for Example 2 is expressed as

$$\mathbb{U}(\varphi, \mathfrak{F}) = \mathbb{U}_{0}(\varphi, \mathfrak{F}) + \mathbb{U}_{1}(\varphi, \mathfrak{F}) + \mathbb{U}_{2}(\varphi, \mathfrak{F}) + \cdots \mathbb{U}_{j}(\varphi, \mathfrak{F}),$$

$$\mathbb{V}(\varphi, \mathfrak{F}) = \mathbb{V}_{0}(\varphi, \mathfrak{F}) + \mathbb{V}_{1}(\varphi, \mathfrak{F}) + \mathbb{V}_{2}(\varphi, \mathfrak{F}) + \cdots \mathbb{V}_{j}(\varphi, \mathfrak{F}).$$
(65)

By putting $\gamma = 1$, we obtain the actual result of the system of Korteweg–de Vries equation (??):

$$\mathbb{U}(\varphi, \mathfrak{F}) = \rho \left(\tanh\left(\frac{\varrho}{2} + \frac{\rho\varphi}{2} - \frac{\rho^2 \mathfrak{F}}{2}\right) + 1 \right),$$

$$\mathbb{V}(\varphi, \mathfrak{F}) = \frac{\rho^2}{2} \sec h^2 \left(\frac{\varrho}{2} + \frac{\rho\varphi}{2} - \frac{\rho^2 \mathfrak{F}}{2}\right) - 1.$$
(67)

In Figures 5 and 6, the actual and analytical solutions of $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$ are proved at $\delta = 2, \rho = 0.5$, and $\varrho = 1$. In Figures 7 and 8, the surface and two-dimensional figure for $\mathbb{U}(\varphi, \mathfrak{F})$ and $\mathbb{V}(\varphi, \mathfrak{F})$ for numerous fractional orders are described which demonstrate that the modified decomposition technique and new iterative transformation technique approximated obtained results are in close contact with the analytical and the exact results. This comparison shows a strong connection among the modified decomposition method and actual solutions. Consequently, the modified decomposition technique and new iterative transformation technique decomposition technique and new iterative transformation for $\mathbb{C}(\varphi, \mathfrak{F})$. technique are accurate innovative techniques which need less calculation time and are very simple and more flexible than the homotopy analysis technique and homotopy perturbation technique.

Example 3. (see [31]). Consider the fractional-order nonlinear system of modified Korteweg-de Vries equations given as (2) with the conditions

$$\mathbb{U}(\varphi, 0) = \frac{2 + \tanh \varphi}{2},$$

$$\mathbb{V}(\varphi, 0) = \frac{2 - \tanh \varphi}{4},$$

$$\mathbb{W}(\varphi, 0) = 2 - \tanh \varphi.$$
(68)

Case I: first, we apply the modified decomposition technique for Example 3.

Using the Laplace transformation to (2), we have

$$\frac{\nu^{\gamma}}{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}\mathcal{U}\left(\varphi,\nu\right)-\sum_{\kappa=0}^{j-1}\left(\frac{1}{\nu}\right)^{\gamma-\kappa-1}\mathbb{U}^{(\kappa)}\left(0\right)=\mathbb{E}\left[\frac{1}{2}\frac{\partial^{3}\mathbb{U}}{\partial\mathfrak{F}^{3}}-3\mathbb{U}^{2}\frac{\partial\mathbb{U}}{\partial\varphi}+\frac{3}{2}\mathbb{W}\frac{\partial^{2}\mathbb{V}}{\partial\varphi^{2}}+3\frac{\partial\mathbb{W}}{\partial\varphi}\frac{\partial\mathbb{W}}{\partial\varphi}+\frac{3}{2}\mathbb{V}\frac{\partial^{2}\mathbb{W}}{\partial\varphi^{2}}\right]$$
$$+3yx\frac{\partial\mathbb{U}}{\partial\varphi}+3zx\frac{\partial\mathbb{V}}{\partial\varphi}+3zy\frac{\partial\mathbb{W}}{\partial\varphi}\right],$$

$$\frac{v^{\gamma}}{(v^{\gamma}(1-\gamma)+\gamma)}\mathcal{V}(\varphi,v) - \sum_{\kappa=0}^{j-1} \left(\frac{1}{v}\right)^{\gamma-\kappa-1} \mathbb{V}^{(\kappa)}(0) = \mathbb{L}\left[-\frac{\partial^{3}\mathbb{V}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi}\frac{\partial\mathbb{V}}{\partial\varphi} - 3\mathbb{V}\frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{V}^{2}\frac{\partial\mathbb{W}}{\partial\varphi} + 6zy\frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2}\frac{\partial\mathbb{V}}{\partial\varphi}\right],$$

$$\frac{v^{\gamma}}{(v^{\gamma}(1-\gamma)+\gamma)}\mathcal{W}(\varphi,v) - \sum_{\kappa=0}^{j-1} \left(\frac{1}{v}\right)^{\gamma-\kappa-1} \mathbb{W}^{(\kappa)}(0) = \mathbb{L}\left[-\frac{\partial^{3}\mathbb{W}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi}\frac{\partial\mathbb{W}}{\partial\varphi} - 3\mathbb{W}\frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{V}^{2}\frac{\partial\mathbb{W}}{\partial\varphi} + 6zy\frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2}\frac{\partial\mathbb{V}}{\partial\varphi}\right].$$

$$(69)$$

In view of (68) and straightforward calculations,

$$\mathscr{U}(\varphi,\nu) = \frac{1}{\nu} \mathbb{U}^{(0)}(\varphi,0) + \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}} \mathbb{L}\left[\frac{1}{2}\frac{\partial^{3}\mathbb{U}}{\partial\mathfrak{T}^{3}} - 3\mathbb{U}^{2}\frac{\partial\mathbb{U}}{\partial\varphi} + \frac{3}{2}\mathbb{W}\frac{\partial^{2}\mathbb{W}}{\partial\varphi^{2}} + 3\frac{\partial\mathbb{W}}{\partial\varphi}\frac{\partial\mathbb{W}}{\partial\varphi} + \frac{3}{2}\mathbb{W}\frac{\partial^{2}\mathbb{W}}{\partial\varphi^{2}} + 3yx\frac{\partial\mathbb{W}}{\partial\varphi^{2}} + 3yx\frac{\partial\mathbb{W}}{\partial\varphi} + 3zy\frac{\partial\mathbb{W}}{\partial\varphi}\right],$$

$$\mathscr{V}(\varphi,\nu) = \frac{1}{\nu}\mathbb{V}^{(0)}(\varphi,0) + \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[-\frac{\partial^{3}\mathbb{W}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi}\frac{\partial\mathbb{W}}{\partial\varphi} - 3\mathbb{W}\frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{W}^{2}\frac{\partial\mathbb{W}}{\partial\varphi} + 6zy\frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2}\frac{\partial\mathbb{W}}{\partial\varphi}\right],$$

$$\mathscr{W}(\varphi,\nu) = \frac{1}{\nu}\mathbb{W}^{(0)}(\varphi,0) + \frac{\left(\nu^{\gamma}\left(1-\gamma\right)+\gamma\right)}{\nu^{\gamma}}\mathbb{L}\left[-\frac{\partial^{3}\mathbb{W}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi}\frac{\partial\mathbb{W}}{\partial\varphi} - 3\mathbb{W}\frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{W}^{2}\frac{\partial\mathbb{U}}{\partial\varphi} + 6zx\frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2}\frac{\partial\mathbb{W}}{\partial\varphi}\right].$$
(70)

Applying the Laplace transform, we have

$$\begin{split} & \mathbb{U}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{U}(\varphi,0) \Big] + \mathbb{L}^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \Big[\frac{1}{2} \frac{\partial^{3}\mathbb{U}}{\partial\mathfrak{F}^{3}} - 3\mathbb{U}^{2} \frac{\partial\mathbb{U}}{\partial\varphi} + \frac{3}{2} \mathbb{W} \frac{\partial^{2}\mathbb{W}}{\partial\varphi^{2}} + 3\frac{\partial\mathbb{W}}{\partial\varphi} \frac{\partial\mathbb{W}}{\partial\varphi} + \frac{3}{2} \mathbb{V} \frac{\partial^{2}\mathbb{W}}{\partial\varphi^{2}} + 3yx \frac{\partial\mathbb{U}}{\partial\varphi} \\ & + 3zx \frac{\partial\mathbb{V}}{\partial\varphi} + 3zy \frac{\partial\mathbb{W}}{\partial\varphi} \Big] \Big], \\ & \mathbb{V}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{V}(\varphi,0) \Big] + \mathbb{L}^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \Big[\frac{\partial^{3}\mathbb{V}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi} \frac{\partial\mathbb{V}}{\partial\varphi} - 3\mathbb{V} \frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{V}^{2} \frac{\partial\mathbb{W}}{\partial\varphi} + 6zy \frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2} \frac{\partial\mathbb{V}}{\partial\varphi} \Big] \Big], \\ & \mathbb{W}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1} \Big[\frac{1}{\nu} \mathbb{W}(\varphi,0) \Big] + \mathbb{L}^{-1} \Big[\frac{(\nu^{\gamma}(1-\gamma)+\gamma)}{\nu^{\gamma}} \mathbb{L} \Big[\frac{\partial^{3}\mathbb{W}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}}{\partial\varphi} \frac{\partial\mathbb{W}}{\partial\varphi} - 3\mathbb{W} \frac{\partial^{2}\mathbb{U}}{\partial\varphi^{2}} - 3\mathbb{W}^{2} \frac{\partial\mathbb{W}}{\partial\varphi} + 6zx \frac{\partial\mathbb{U}}{\partial\varphi} + 3\mathbb{U}^{2} \frac{\partial\mathbb{W}}{\partial\varphi} \Big] \Big]. \end{split}$$

$$(71)$$

By the consequence of the modified decomposition technique, we get

$$\mathbb{U}_{0}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{1}{\nu} \mathbb{U}(\varphi, 0) \right] = \frac{1}{2} \mathbb{L}^{-1} \left[\frac{1}{\nu} (2 + \tanh \varphi) \right]$$

$$\mathbb{V}_{0}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{1}{\nu} \mathbb{V}(\varphi, 0) \right] = \frac{1}{4} (2 - \tanh \varphi),$$

$$\mathbb{W}_{0}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{1}{\nu} \mathbb{W}(\varphi, 0) \right] = (2 - \tanh \varphi).$$
(72)

It follows that

$$\sum_{j=0}^{\infty} \mathbb{U}_{j+1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \left[\frac{1}{2} \sum_{j=0}^{\infty} \left(\mathbb{U}_{\varphi \varphi \varphi} \right)_{j} - 3 \sum_{j=0}^{\infty} \mathscr{E}_{j} + \frac{3}{2} \sum_{j=0}^{\infty} \mathscr{E}_{j} + 3 \sum_{j=0}^{\infty} \mathscr{E}_{j} + \frac{3}{2} \sum_{j=0}^{\infty} \mathscr{E}_{j} \right] \right], \quad j$$

$$+3 \sum_{j=0}^{\infty} I_{j} + 3 \sum_{j=0}^{\infty} \mathscr{E}_{j} + 3 \sum_{j=0}^{\infty} \mathscr{E}_{j} \right], \quad j$$

$$\sum_{j=0}^{\infty} \mathbb{V}_{j+1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \left[-\sum_{j=0}^{\infty} \left(\mathbb{V}_{\varphi \varphi \varphi} \right)_{j} - 3 \sum_{j=0}^{\infty} \mathscr{M}_{j} - 3 \sum_{j=0}^{\infty} \mathscr{M}_{j} - 3 \sum_{j=0}^{\infty} \mathscr{D}_{j} + 6 \sum_{j=0}^{\infty} x_{j} + 3 \sum_{j=0}^{\infty} \mathscr{Q}_{j} \right] \right], \quad (73)$$

$$\sum_{j=0}^{\infty} \mathbb{W}_{j+1}(\varphi, \mathfrak{F}) = \mathbb{L}^{-1} \left[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \left[-\sum_{j=0}^{\infty} \left(\mathbb{W}_{\varphi \varphi \varphi} \right)_{j} - 3 \sum_{j=0}^{\infty} \mathscr{R}_{j} - 3 \sum_{j=0}^{\infty} \mathscr{W}_{j} - 3 \sum_{j=0}^{\infty} \mathscr{F}_{j} + 6 \sum_{j=0}^{\infty} \mathscr{X}_{j} + 3 \sum_{j=0}^{\infty} \mathscr{Y}_{j} \right] \right].$$

The Adomian polynomials' some terms are defined as

$$\begin{split} \mathscr{C}_{J} \Big(\mathbb{U}^{2} \mathbb{U}_{\varphi} \Big) &= \begin{cases} \mathbb{U}_{0}^{2} \mathbb{U}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{U}_{0}\mathbb{U}_{1})\mathbb{U}_{0\varphi} + \mathbb{U}_{0}^{2}\mathbb{U}_{1\varphi}, & \text{for } J = 1, \\ (2\mathbb{U}_{0}\mathbb{U}_{2} + \mathbb{U}_{1}^{2})\mathbb{U}_{0\varphi} + (2\mathbb{U}_{0}\mathbb{U}_{1})\mathbb{U}_{1\varphi} + \mathbb{U}_{0}^{2}\mathbb{U}_{2\varphi}, & \text{for } J = 2, \end{cases} \\ \mathscr{F}_{J} \Big(\mathbb{W}\mathbb{V}_{\varphi\varphi} \Big) &= \begin{cases} \mathbb{W}_{0}\mathbb{V}_{0\varphi\varphi}, & \text{for } J = 0, \\ \mathbb{W}_{1}\mathbb{V}_{0\varphi\varphi} + \mathbb{W}_{0}\mathbb{V}_{1\varphi\varphi}, & \text{for } J = 1, \\ \mathbb{W}_{2}\mathbb{V}_{0\varphi\varphi} + \mathbb{W}_{1}\mathbb{V}_{1\varphi\varphi} + \mathbb{W}_{0}\mathbb{V}_{2\varphi\varphi}, & \text{for } J = 2, \end{cases} \\ \mathscr{F}_{J} \Big(\mathbb{W}_{\varphi} \mathbb{W}_{\varphi} \Big) &= \begin{cases} \mathbb{V}_{0\varphi}\mathbb{W}_{0\varphi}, & \text{for } J = 0, \\ \mathbb{V}_{0\varphi}\mathbb{W}_{1\varphi\varphi} + \mathbb{V}_{1\varphi}\mathbb{W}_{0\varphi}, & \text{for } J = 0, \\ \mathbb{V}_{2\varphi}\mathbb{W}_{0\varphi\varphi}, & \text{for } J = 0, \end{cases} \\ \mathbb{V}_{0\varphi}\mathbb{W}_{1\varphi\varphi} + \mathbb{V}_{1\varphi}\mathbb{W}_{1\varphi\varphi} + \mathbb{V}_{0\varphi}\mathbb{W}_{2\varphi\varphi}, & \text{for } J = 2, \end{cases} \end{cases} \\ \mathscr{F}_{J} \Big(\mathbb{W}_{Z}\mathbb{U}_{\varphi} \Big) &= \begin{cases} (y_{0\varphi}\mathbb{W}_{0\varphi\varphi}, & \text{for } J = 0, \\ \mathbb{V}_{0\varphi}\mathbb{W}_{0\varphi\varphi}, & \text{for } J = 0, \end{cases} \\ \mathbb{V}_{2\varphi}\mathbb{W}_{0\varphi\varphi} + \mathbb{V}_{1\varphi}\mathbb{W}_{1\varphi\varphi}, & \text{for } J = 1, \end{cases} \\ \mathbb{V}_{2\varphi}\mathbb{W}_{0\varphi\varphi} + \mathbb{V}_{1\varphi}\mathbb{W}_{1\varphi\varphi}, & \text{for } J = 1, \end{cases} \\ \mathbb{V}_{2\varphi}\mathbb{W}_{0\varphi\varphi} + \mathbb{V}_{1\varphi}\mathbb{W}_{1\varphi\varphi}, & \text{for } J = 1, \end{cases} \\ \mathbb{V}_{2\varphi}\mathbb{W}_{0\varphi\varphi} + (y_{1}\mathbb{U}_{1\varphi}, & \text{for } J = 1, \end{cases} \\ \mathbb{V}_{0}\mathbb{U}_{0}\mathbb{U}_{1\varphi} + (y_{1}\mathbb{U}_{0\varphi}, & \text{for } J = 1, \end{cases} \\ (y_{1}\mathbb{U}_{0}\mathbb{U}_{\varphi}) &= \begin{cases} (z_{1}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 1, \\ (z_{2}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 0, \end{cases} \\ (z_{2}\mathbb{U}_{0}\mathbb{U}_{1\varphi} + (z_{2}\mathbb{U}_{1}\mathbb{U}_{0\varphi}, & \text{for } J = 2, \end{cases} \\ \mathscr{F}_{J} \Big(\mathbb{U}_{Z}\mathbb{W}_{\varphi} \Big) &= \begin{cases} (z_{1}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 1, \\ (z_{2}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 0, \end{cases} \\ (z_{2}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 0, \end{cases} \\ (z_{2}\mathbb{U}_{0}\mathbb{U}_{0\varphi}, & \text{for } J = 0, \end{cases} \\ \mathbb{U}_{0\varphi}\mathbb{V}_{0\varphi} + (z_{2}\mathbb{U}_{1}\mathbb{U}_{1\varphi} + (z_{2}\mathbb{U}_{2}\mathbb{U}_{0\varphi}, & \text{for } J = 2, \end{cases} \\ \mathscr{M}_{J} \Big(\mathbb{U}_{\varphi}\mathbb{W}_{\varphi} \Big) &= \begin{cases} \mathbb{U}_{0}\mathbb{U}_{\varphi}\mathbb{U}_{\varphi} + \mathbb{U}_{\varphi}\mathbb{U}_{\varphi}, & \text{for } J = 1, \\ \mathbb{U}_{2}\mathbb{U}_{\varphi}\mathbb{U}_{\varphi} + \mathbb{U}_{\varphi}\mathbb{U}_{\varphi}, & \text{for } J = 1, \\ \mathbb{U}_{2}\mathbb{U}_{\varphi}\mathbb{U}_{\varphi} + \mathbb{U}_{\varphi}\mathbb{U}_{\varphi}, & \text{for } J = 1, \\ \mathbb{U}_{2}\mathbb{U}_{\varphi}\mathbb{U}_{\varphi} + \mathbb{U}_{\varphi}\mathbb{U}_{\varphi}, & \text{for } J = 2, \end{cases} \end{aligned}$$

$$\begin{split} \mathscr{N}_{I} \Big(\mathbb{VU}_{\varphi \varphi} \Big) &= \begin{cases} \mathbb{V}_{0} \mathbb{U}_{0\varphi \varphi}, & \text{for } J = 0, \\ \mathbb{V}_{0} \mathbb{U}_{1\varphi \varphi} + \mathbb{V}_{1} \mathbb{U}_{0\varphi \varphi}, & \text{for } J = 1, \\ \mathbb{V}_{2} \mathbb{U}_{0\varphi \varphi} + \mathbb{V}_{1} \mathbb{U}_{1\varphi \varphi} + \mathbb{V}_{0} \mathbb{U}_{2\varphi \varphi}, & \text{for } J = 2, \end{cases} \\ \\ \mathscr{O}_{J} \Big(\mathbb{V}^{2} \mathbb{W}_{\varphi} \Big) &= \begin{cases} \mathbb{V}_{0}^{2} \mathbb{W}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{V}_{0} \mathbb{V}_{1}) \mathbb{W}_{0\varphi} + (2\mathbb{V}_{0} \mathbb{V}_{1}) \mathbb{W}_{1\varphi} + \mathbb{V}_{0}^{2} \mathbb{W}_{2\varphi}, & \text{for } J = 2, \end{cases} \\ \\ (2\mathbb{V}_{0} \mathbb{V}_{2} + \mathbb{V}_{1}^{2}) \mathbb{U}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{V}_{0} \mathbb{U}_{1\varphi} + (2\mathbb{V}_{1}) \mathbb{U}_{1\varphi} + (2\mathbb{V}_{0} \mathbb{V}_{1}) \mathbb{W}_{1\varphi} + \mathbb{V}_{0}^{2} \mathbb{W}_{2\varphi}, & \text{for } J = 2, \end{cases} \\ \\ \mathscr{A}_{J} \Big(\mathbb{U}^{2} \mathbb{V}_{\varphi} \Big) &= \begin{cases} \mathbb{U}_{0}^{2} \mathbb{V}_{0\varphi}, & \text{for } J = 1, \\ (2\mathbb{U}_{0} \mathbb{U}_{1} + (2\mathbb{V}_{1}) \mathbb{U}_{1\varphi} + (2\mathbb{V}_{2}) \mathbb{U}_{0\varphi}, & \text{for } J = 2, \end{cases} \\ \\ (2\mathbb{U}_{0} \mathbb{U}_{2} + \mathbb{U}_{1}^{2}) \mathbb{V}_{0\varphi} + (2\mathbb{U}_{0} \mathbb{U}_{1}) \mathbb{V}_{1\varphi} + \mathbb{U}_{0}^{2} \mathbb{V}_{2\varphi} & \text{for } J = 2, \end{cases} \\ \\ \\ \mathscr{B}_{0} \Big(\mathbb{U}_{\varphi} \mathbb{V}_{\varphi} \Big) &= \begin{cases} \mathbb{U}_{0\varphi} \mathbb{V}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{U}_{0} \mathbb{U}_{1} + \mathbb{U}_{1\varphi} \mathbb{V}_{0\varphi}, & \text{for } J = 1, \\ (2\mathbb{U}_{0} \mathbb{U}_{2} + \mathbb{U}_{1}^{2}) \mathbb{V}_{0\varphi} + (2\mathbb{U}_{0} \mathbb{U}_{1}) \mathbb{V}_{1\varphi} + \mathbb{U}_{0}^{2} \mathbb{V}_{2\varphi} & \text{for } J = 2, \end{cases} \\ \\ \\ \mathscr{S}_{J} \Big(\mathbb{W} \mathbb{U}_{\varphi\varphi} \Big) &= \begin{cases} \mathbb{W}_{0} \mathbb{U}_{\varphi\varphi}, & \text{for } J = 1, \\ \mathbb{U}_{2\varphi} \mathbb{V}_{0\varphi}, & \text{for } J = 0, \\ \mathbb{U}_{2\varphi} \mathbb{V}_{0\varphi} + \mathbb{U}_{1} \mathbb{U}_{0\varphi\varphi}, & \text{for } J = 2, \end{cases} \\ \\ \\ \mathscr{W}_{0} \mathbb{U}_{1\varphi\varphi} = \mathbb{W}_{1} \mathbb{U}_{1\varphi\varphi} + \mathbb{W}_{1} \mathbb{U}_{0\varphi\varphi}, & \text{for } J = 1, \\ (2\mathbb{W}_{0} \mathbb{W}_{1}) \mathbb{V}_{0\varphi} + (2\mathbb{W}_{0} \mathbb{W}_{1}) \mathbb{V}_{1\varphi}, & \text{for } J = 2, \end{cases} \\ \\ \\ \mathscr{F}_{J} \Big(\mathbb{U} \mathbb{W}_{1} \mathbb{U}_{\varphi} \Big) = \begin{cases} \mathbb{W}_{0}^{2} \mathbb{U}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{U}_{0} \mathbb{U}_{1}) \mathbb{W}_{0\varphi} + (2\mathbb{U}_{0} \mathbb{W}_{1}) \mathbb{V}_{1\varphi}, & \text{for } J = 2, \end{cases} \\ \\ \\ \mathscr{F}_{J} \Big(\mathbb{U}^{2} \mathbb{W}_{\varphi} \Big) = \begin{cases} \mathbb{U}_{0}^{2} \mathbb{W}_{0\varphi}, & \text{for } J = 0, \\ (2\mathbb{U}_{0} \mathbb{U}_{1}) \mathbb{W}_{0\varphi}, & \text{for } J = 1, \\ (2\mathbb{U}_{0} \mathbb{U}_{2} \mathbb{W}_{1}) \mathbb{W}_{0\varphi}, & \text{for } J = 2, \end{cases} \\ \\ \\ \\ \\ \mathscr{F}_{J} \Big(\mathbb{U}^{2} \mathbb{W}_{\varphi} \Big) = \begin{cases} \mathbb{U}_{0}^{2} \mathbb{W}_{0} \mathbb{W}_{0\varphi}, & \text{for } J = 1, \\ (2\mathbb{U}_{0} \mathbb{U}_{2} + \mathbb{U}_{0}^{2}$$

For $j = 0, 1, 2, 3, \ldots$,

$$\begin{split} \mathbb{U}_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \bigg[\frac{1}{2} \Big(\mathbb{U}_{\varphi \varphi \varphi} \Big)_{0} - 3\mathscr{E}_{0} + \frac{3}{2} \mathscr{F}_{0} + 3\mathscr{E}_{0} + \frac{3}{2} \mathscr{H}_{0} + 3I_{0} + 3\mathscr{F}_{0} + 3\mathscr{H}_{0} \bigg] \bigg] \\ &= \frac{11}{2} \sec h^{2}(\varphi) \bigg((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \bigg) \\ \mathbb{V}_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \bigg[\frac{\left(\nu^{\gamma} \left(1 - \gamma \right) + \gamma \right)}{\nu^{\gamma}} \mathbb{L} \big[\left(\mathbb{V}_{\varphi \varphi \varphi} \right)_{0} - 3\mathscr{M}_{0} - 3\mathscr{N}_{0} - 3\mathscr{O}_{0} + 6x_{0} + 3\mathscr{Q}_{0} \big] \bigg] \\ &= -\frac{11}{8} \sec h^{2}(\varphi) \bigg((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \bigg) \end{split}$$

$$\begin{split} \mathbb{W}_{1}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[(\mathbb{W}_{\varphi\varphi\varphi})_{0} - 3\mathcal{R}_{0} - 3\widehat{S}_{0} - 3\mathcal{F}_{0} + 6\mathcal{X}_{0} + 3\mathcal{Y}_{0} \right] \right] \\ &= \frac{11}{2} \sec h^{2}(\varphi) \left((1-\gamma) + \frac{\gamma \mathcal{F}^{y}}{\Gamma(\gamma+1)} \right) \\ \mathbb{U}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[\frac{1}{2} (\mathbb{U}_{\varphi\varphi\varphi})_{1} - 3\mathcal{E}_{1} + \frac{3}{2} \mathcal{F}_{1} + 3\mathcal{F}_{1} + \frac{3}{2} \mathcal{F}_{1} + 3\mathcal{F}_{1} + 3\mathcal{F}_{1} + 3\mathcal{F}_{1} + 3\mathcal{F}_{1} \right] \\ &= \frac{-121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1-\gamma)^{2} + \frac{\gamma^{2} \mathcal{G}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma \mathcal{F}^{y}}{\Gamma(\gamma+1)} \right) \\ \mathbb{V}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[(\mathbb{W}_{\varphi\varphi\varphi})_{1} - 3\mathcal{H}_{1} - 3\mathcal{H}_{1} - 3\mathcal{H}_{1} - 3\mathcal{H}_{1} + 3\mathcal{F}_{1} + 3\mathcal{F}_{1} \right] \\ &= \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1-\gamma)^{2} + \frac{\gamma^{2} \mathcal{G}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma \mathcal{F}^{y}}{\Gamma(2\gamma+1)} \right) \\ \mathbb{W}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[(\mathbb{W}_{\varphi\varphi\varphi})_{1} - 3\mathcal{H}_{1} - 3\mathcal{F}_{1} - 3\mathcal{F}_{1} + 6\mathcal{H}_{1} + 3\mathcal{F}_{1} \right] \\ &= \frac{242}{8} anh(\varphi) \sec h^{2}(\varphi) \left((1-\gamma)^{2} + \frac{\gamma^{2} \mathcal{G}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma \mathcal{F}^{y}}{\Gamma(2\gamma+1)} \right) \\ \mathbb{U}_{3}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[\frac{1}{2} (\mathbb{U}_{\varphi\varphi\varphi\varphi})_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} + 3\mathcal{H}_{2} + 3\mathcal{H}_{2} + 3\mathcal{F}_{2} + 3\mathcal{H}_{2} \right] \right] \\ &= \frac{1331}{48} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1-\gamma)^{3} + \gamma(1-\gamma)(1+\gamma+2\gamma^{2}) \frac{\mathcal{F}^{y}}{\Gamma(\gamma+1)} + \frac{3\gamma^{2}(1-\gamma)\mathcal{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{\gamma^{3}\Gamma(2\gamma+1)\mathcal{F}^{3\gamma}}{\Gamma(3\gamma+1)} \right\} \\ \mathbb{V}_{3}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[(\mathbb{W}_{\varphi\varphi\varphi\varphi})_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} + 3\mathcal{H}_{2} + 3\mathcal{H}_{2} \right] \right] \\ &= \frac{2662}{96} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1-\gamma)^{3} + \gamma(1-\gamma)(1+\gamma+2\gamma^{2}) \frac{\mathcal{F}^{y}}{\Gamma(\gamma+1)} + \frac{3\gamma^{2}(1-\gamma)\mathcal{F}^{2\gamma}}{\Gamma(2\gamma+1)} + \frac{\gamma^{3}\Gamma(2\gamma+1)\mathcal{F}^{3\gamma}}{\Gamma(3\gamma+1)} \right\} \\ \mathbb{W}_{3}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{y}(1-\gamma)+\gamma)}{\nu^{y}} \mathbb{L} \left[(\mathbb{W}_{\varphi\varphi\varphi\varphi})_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} - 3\mathcal{H}_{2} + 3\mathcal{H}_{2} + 3\mathcal{H}_{2} + 3\mathcal{H}_{2} \right] \right] \\ &= \frac{2662}{96} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1-\gamma)^{3} + \gamma(1-\gamma)(1+\gamma+2\gamma^{2}) \frac{\mathcal{F}^{y}}{\Gamma(\gamma+1)} + \frac{3\gamma^{2}(1-\gamma)\mathcal{F}^{2\gamma}}{$$

The modified decomposition technique result for Example 3 is given as

$$\begin{split} \mathbb{U}(\varphi,\mathfrak{F}) &= \mathbb{U}_{0}(\varphi,\mathfrak{F}) + \mathbb{U}_{1}(\varphi,\mathfrak{F}) + \mathbb{U}_{2}(\varphi,\mathfrak{F}) + \mathbb{U}_{3}(\varphi,\mathfrak{F}) \dots, \\ &= \frac{1}{2}\left(2 + \tanh \varphi\right) + \frac{11}{2} \sec h^{2}(\varphi) \left(\left(1 - \gamma\right) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)}\right) \\ &- \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left(\left(1 - \gamma\right)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)}\right) \\ &+ \frac{1331}{48} \sec h^{4}(\varphi) \left[\cosh(2\varphi) - 2\right] \left\{ \left(1 - \gamma\right)^{3} + \gamma \left(1 - \gamma\right) \left(1 + \gamma + 2\gamma^{2}\right) \frac{\mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} \\ &+ \cdots . \end{split}$$

(76)



FIGURE 1: The actual and analytical (MDM/NITM) result figure at $U(\varphi, \mathfrak{F})$ of Example 1 for $\varrho = 1$, $\rho = 0.5$, and $\delta = 2$.



FIGURE 2: The actual and analytical (MDM/NITM) result figure at $\mathbb{V}(\varphi, \mathfrak{F})$ of Example 1 for $\varrho = 1$, $\rho = 0.5$, and $\delta = 2$. Consequently, we get

$$\mathbb{V}(\varphi, \mathfrak{F}) = \frac{1}{4} (2 - \tanh\varphi) - \frac{11}{8} \sec h^{2}(\varphi) \left((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) + \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) \\ - \frac{1331}{48} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1 - \gamma)^{3} + \gamma(1 - \gamma)(1 + \gamma + 2\gamma^{2}) \frac{\mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} + \cdots$$

$$\mathbb{W}(\varphi, \mathfrak{F}) = (2 - \tanh\varphi) - \frac{11}{2} \sec h^{2}(\varphi) \left((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) + \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1 - \gamma)^{2} + \frac{\gamma^{2}\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) \\ - \frac{2662}{96} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1 - \gamma)^{3} + \gamma(1 - \gamma)(1 + \gamma + 2\gamma^{2}) \frac{\mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} + \cdots$$

$$(77)$$



FIGURE 3: Analytical investigation of figure $\mathbb{U}(\varphi, \mathfrak{T})$ for Example 1 for different fractional orders $\gamma = 1.0, 0.8, 0.6, 0.4, \rho = 0.5, \varrho = 1$, and $\delta = 2$.



FIGURE 4: Analytical investigation of figure $\mathbb{V}(\varphi, \mathfrak{F})$ for Example 1 for different fractional orders $\gamma = 1.0, 0.8, 0.6, 0.4, \rho = 0.5, \varrho = 1$, and $\delta = 2$.



FIGURE 5: The analytical and exact (MDM/NITM) solution plot at $U(\varphi, \mathfrak{F})$ of Example 2 for $\rho = 0.5$, $\varrho = 1$, and $\delta = 2$.



FIGURE 6: The analytical and exact (MDM/NITM) solution plot at $V(\varphi, \mathfrak{F})$ of Example 2 for $\rho = 0.5$, $\varrho = 1$, and $\delta = 2$.

By putting $\gamma = 1$, we obtain the exact result of the system of Korteweg–de Vries equation (2):

$$\mathbb{U}(\varphi, \mathfrak{F}) = \frac{1}{2} \left(2 + \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right),$$

$$\mathbb{V}(\varphi, \mathfrak{F}) = \frac{1}{4} \left(2 - \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right),$$

$$\mathbb{W}(\varphi, \mathfrak{F}) = \left(2 - \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right).$$
(78)



FIGURE 7: Mathematical analysis of the plot of $\mathbb{V}(\varphi, \mathfrak{F})$ for Example 2 for different fractional orders $\gamma = 1.0, 0.8, 0.6, 0.4, \rho = 0.5, \varrho = 1$, and $\delta = 2$.



FIGURE 8: Mathematical analysis of the plot of $\mathbb{V}(\varphi, \mathfrak{F})$ for Example 2 for different fractional orders $\gamma = 1.0, 0.8, 0.6, 0.4\rho = 0.5, \varrho = 1$, and $\delta = 2$.



FIGURE 9: The analytical and exact result plot at $U(\varphi, \mathfrak{F})$ of Example 3 for $\rho = 0.5$, $\varrho = 1$, and $\delta = 2$.

Case II: now, we apply the new iterative transformation technique for Example 3.

By using the suggested analytical method, we get

$$\begin{split} & \mathbb{U}_{0}(\varphi,\mathfrak{F}) = \frac{1}{2}\left(2 + \tanh\varphi\right) \\ & \mathbb{V}_{0}(\varphi,\mathfrak{F}) = \frac{1}{4}\left(2 - \tanh\varphi\right) \\ & \mathbb{W}_{0}(\varphi,\mathfrak{F}) = \left(2 - \tanh\varphi\right) \\ & \mathbb{U}_{1}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}\left(1 - \gamma\right) + \gamma\right)}{\nu^{\gamma}}\mathbb{U}\left[\frac{1}{2}\frac{\partial^{3}\mathbb{U}_{0}}{\partial\mathfrak{F}^{3}} - 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + \frac{3}{2}\mathbb{W}_{0}\frac{\partial^{2}\mathbb{W}_{0}}{\partial\varphi} + \frac{3}{2}\mathbb{V}_{0}\frac{\partial^{2}\mathbb{W}_{0}}{\partial\varphi^{2}} + 3\mathbb{V}_{0}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{0}}{\partial\varphi}\right]\right] \\ & = \frac{11}{2}\sec h^{2}(\varphi)\mathbb{L}^{-1}\left[\frac{\omega^{\nu'1}}{\nu^{\nu'2}}\right] \\ & \mathbb{V}_{1}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}\left(1 - \gamma\right) + \gamma\right)}{\nu^{\gamma}}\mathbb{U}\left[\frac{\partial^{3}\mathbb{W}_{0}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}_{0}}{\partial\varphi}\frac{\partial\mathbb{W}_{0}}{\partial\varphi} - 3\mathbb{W}_{0}\frac{\partial^{2}\mathbb{U}_{0}}{\partial\varphi^{2}} - 3\mathbb{V}_{0}^{2}\frac{\partial\mathbb{W}_{0}}{\partial\varphi} + 6\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{W}_{0}}{\partial\varphi}\right]\right] \\ & = -\frac{11}{8}\sec h^{2}(\varphi)\left(\left(1 - \gamma\right) + \frac{\gamma\mathfrak{S}^{\gamma}}{\Gamma(\gamma + 1)}\right) \\ & \mathbb{W}_{1}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}\left(1 - \gamma\right) + \gamma\right)}{\nu^{\gamma}}\mathbb{U}\left[\frac{\partial^{3}\mathbb{W}_{0}}{\partial\varphi^{3}} - 3\frac{\partial\mathbb{U}_{0}}{\partial\varphi}\frac{\partial\mathbb{W}_{0}}{\partial\varphi} - 3\mathbb{W}_{0}\frac{\partial^{2}\mathbb{U}_{0}}{\partial\varphi^{2}} - 3\mathbb{W}_{0}^{2}\frac{\partial\mathbb{W}_{0}}{\partial\varphi} + 6\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{U}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{W}_{0}}{\partial\varphi}\right]\right] \\ & = -\frac{11}{8}\sec h^{2}(\varphi)\left(\left(1 - \gamma\right) + \frac{\gamma\mathfrak{S}^{\gamma}}{\Gamma(\gamma + 1)}\right) \\ & \mathbb{U}_{2}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}\left(1 - \gamma\right) + \gamma\right)}{\nu^{\gamma}}\mathbb{U}\left[\frac{1}{2}\frac{\partial^{3}\mathbb{U}_{1}}{\partial\mathfrak{F}^{3}} - 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{U}_{1}}{\partial\varphi} + \frac{3}{2}\mathbb{W}_{0}\frac{\partial^{2}\mathbb{W}_{1}}{\partial\varphi} + 3\frac{\partial\mathbb{W}_{0}}{\partial\varphi} + 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{W}_{0}}{\partial\varphi}\right]\right] \\ & = -\frac{11}{2}\sec h^{2}(\varphi)\left(\left(1 - \gamma\right) + \frac{\gamma\mathfrak{S}^{\gamma}}{\Gamma(\gamma + 1)}\right) \\ & \mathbb{U}_{2}(\varphi,\mathfrak{F}) = \mathbb{L}^{-1}\left[\frac{\left(\nu^{\gamma}\left(1 - \gamma\right) + \gamma\right)}{\nu^{\gamma}}\mathbb{U}\left[\frac{1}{2}\frac{\partial^{3}\mathbb{U}_{1}}{\partial\mathfrak{F}^{3}} - 3\mathbb{U}_{0}^{2}\frac{\partial\mathbb{U}_{1}}{\partial\varphi} + \frac{3}{2}\mathbb{W}_{0}\frac{\partial^{2}\mathbb{W}_{1}}{\partial\varphi} + \frac{3}{2}\mathbb{W}_{1}\frac{\partial^{2}\mathbb{W}_{1}}{\partial\varphi^{2}} + 3\mathbb{W}_{1}\frac{\partial^{2}\mathbb{W}_{1}}{\partial\varphi^{2}} + 3\mathbb{W}_{1}\mathbb{W}_{1}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi} + 3\mathbb{U}_{0}\mathbb{W}_{0}\frac{\partial\mathbb{W}_{1}}{\partial\varphi$$

$$\begin{split} \mathbb{V}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \left[\frac{\partial^{3} V_{1}}{\partial \varphi^{3}} - 3\frac{\partial U_{1}}{\partial \varphi} \frac{\partial V_{1}}{\partial \varphi} - 3V_{1}^{2} \frac{\partial^{2} U_{1}}{\partial \varphi} - 3V_{1}^{2} \frac{\partial^{2} U_{1}}{\partial \varphi} + 6U_{1}V_{1}^{2} \frac{\partial U_{1}}{\partial \varphi} + 3U_{1}^{2} \frac{\partial V_{1}}{\partial \varphi} \right] \right] \\ &= \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{S}^{2}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma \mathfrak{S}^{2}}{\Gamma(\gamma+1)} \right) \\ \mathbb{W}_{2}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \left[\frac{\partial^{3} U_{1}}{\partial \varphi^{3}} - 3\frac{\partial U_{1}}{\partial \varphi} \frac{\partial W_{1}}{\partial \varphi} - 3W_{1}^{2} \frac{\partial^{2} U_{1}}{\partial \varphi^{2}} - 3W_{1}^{2} \frac{\partial^{2} U_{1}}{\partial \varphi} + 6U_{1}W_{1} \frac{\partial U_{1}}{\partial \varphi} + 3U_{1}^{2} \frac{\partial W_{1}}{\partial \varphi} \right] \right] \\ &= \frac{242}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1-\gamma)^{2} + \frac{\gamma^{2} \mathfrak{S}^{2}}{\Gamma(2\gamma+1)} + \frac{2(1-\gamma)\gamma \mathfrak{S}^{2}}{\Gamma(\gamma+1)} \right) \\ \mathbb{U}_{3}(\varphi,\mathfrak{F}) &= \mathbb{L}^{-1} \left[\frac{(\nu^{2}(1-\gamma)+\gamma)}{\nu^{2}} \mathbb{L} \left[\frac{1}{2} \frac{\partial^{2} U_{2}}{\partial \mathfrak{S}^{3}} - 3U_{2}^{2} \frac{\partial^{2} V_{2}}{\partial \varphi} + \frac{3}{2} W_{2}^{2} \frac{\partial^{2} V_{2}}{\partial \varphi} + \frac{3}{2} V_{2}^{2} \frac{\partial^{2} W_{2}}{\partial \varphi^{2}} + 3V_{2} W_{2}^{2} \frac{\partial U_{2}}{\partial \varphi} + 3U_{2} W_{2}^{2} \frac{\partial V_{2}}{\partial \varphi} + 3U_{2} V_{2}^{2} \frac{\partial W_{2}}{\partial \varphi} + 3U_{2} W_{2}^{2} \frac{\partial U_{2}}{\partial \varphi} + 3U_{2} W_{2}^{2} \frac{\partial V_{2}}{\partial \varphi} + 3U_{2} W_{2} \frac{\partial V_{2}}{\partial \varphi} + 3U$$

The series of results for Example 3 is given as

$$\mathbb{U}(\varphi,\mathfrak{F}) = \mathbb{U}_0(\varphi,\mathfrak{F}) + \mathbb{U}_1(\varphi,\mathfrak{F}) + \mathbb{U}_2(\varphi,\mathfrak{F}) + \mathbb{U}_3(\varphi,\mathfrak{F}) + \cdots \mathbb{U}_j(\varphi,\mathfrak{F})$$

$$= \frac{1}{2} (2 + \tanh \varphi) + \frac{11}{2} \sec h^{2}(\varphi) \left((1 - \gamma) + \frac{\gamma \mathfrak{V}}{\Gamma(\gamma + 1)} \right)$$

$$- \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{V}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{V}}{\Gamma(\gamma + 1)} \right)$$

$$+ \frac{1331}{48} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1 - \gamma)^{3} + \gamma(1 - \gamma) (1 + \gamma + 2\gamma^{2}) \frac{\mathfrak{V}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{V}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{V}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} + \cdots.$$
(80)



FIGURE 10: The analytical and exact result plot at $\mathbb{V}(\varphi, \mathfrak{F})$ of Example 3 for $\rho = 0.5$, $\varrho = 1$, and $\delta = 2$.

Consequently, we get

$$\mathbb{V}(\varphi, \mathfrak{F}) = \frac{1}{4} (2 - \tanh\varphi) - \frac{11}{8} \sec h^{2}(\varphi) \left((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) + \frac{121}{8} \tanh(\varphi) \sec h^{2}(\varphi) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) \\ - \frac{1331}{48} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1 - \gamma)^{3} + \gamma(1 - \gamma)(1 + \gamma + 2\gamma^{2}) \frac{\mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} + \cdots, \\ \mathbb{W}(\varphi, \mathfrak{F}) = (2 - \tanh\varphi) - \frac{11}{2} \sec h^{2}(\varphi) \left((1 - \gamma) + \frac{\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) + \frac{121}{4} \tanh(\varphi) \sec h^{2}(\varphi) \left((1 - \gamma)^{2} + \frac{\gamma^{2} \mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{2(1 - \gamma)\gamma \mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} \right) \\ - \frac{2662}{48} \sec h^{4}(\varphi) [\cosh(2\varphi) - 2] \left\{ (1 - \gamma)^{3} + \gamma(1 - \gamma)(1 + \gamma + 2\gamma^{2}) \frac{\mathfrak{F}^{\gamma}}{\Gamma(\gamma + 1)} + \frac{3\gamma^{2}(1 - \gamma)\mathfrak{F}^{2\gamma}}{\Gamma(2\gamma + 1)} + \frac{\gamma^{3}\Gamma(2\gamma + 1)\mathfrak{F}^{3\gamma}}{\Gamma(3\gamma + 1)} \right\} + \cdots.$$

$$(81)$$

By putting $\gamma = 1$, we obtain the exact result of modified couple Korteweg–de Vries equation (2):

$$\mathbb{U}(\varphi, \mathfrak{F}) = \frac{1}{2} \left(2 + \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right),$$

$$\mathbb{V}(\varphi, \mathfrak{F}) = \frac{1}{4} \left(2 - \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right),$$

$$\mathbb{W}(\varphi, \mathfrak{F}) = \left(2 - \tanh\left(\varphi - \frac{11\mathfrak{F}}{2}\right) \right).$$
(82)

In Figures 9–11, the actual and analytical solutions of $\mathbb{U}(\varphi, \mathfrak{F}), \mathbb{V}(\varphi, \mathfrak{F})$, and $\mathbb{W}(\varphi, \mathfrak{F})$ are proved at $\delta = 2, \rho = 0.5$, and $\varrho = 1$. In Figures 12–14, the surface and two-

dimensional figure for $U(\varphi, \mathfrak{F}), V(\varphi, \mathfrak{F})$, and $W(\varphi, \mathfrak{F})$ for numerous fractional orders are described which demonstrate that the modified decomposition technique and



FIGURE 11: The analytical and exact result plot at $\mathbb{W}(\varphi, \mathfrak{F})$ of Example 3 for $\rho = 0.5$, $\varrho = 1$, and $\delta = 2$.



FIGURE 12: The three and two dimensional different fractional order of Example 3 with respect to $U(\varphi, \mathfrak{F})$.

new iterative transformation technique approximated obtained results are in close contact with the analytical and the exact results. This comparison shows a strong connection among the modified decomposition method and actual solutions. Consequently, the modified decomposition technique and new iterative transformation technique are accurate innovative techniques which need less calculation time and are very simple and more flexible than the homotopy analysis technique and homotopy perturbation technique.



FIGURE 13: The three and two dimensional different fractional order of Example 3 with respect to $V(\varphi, \mathfrak{F})$.



FIGURE 14: The three and two dimensional different fractional order of Example 3 with respect to $W(\varphi, \mathfrak{F})$.
7. Conclusion

In this article, we have considered the nonlinear fractionalorder Korteweg-de Vries equations in the sense of the Atangana-Baleanu derivative which is able to perform more extensive analysis due to the nonsingular kernel in its structure. The mathematical solutions are obtained with the help of the modified decomposition method and new iterative transformation method associated with the Atangana-Baleanu derivative. The present analysis illuminates the effectiveness of the considered derivative operator. We can conclude from the analytical results that these are very reliable, simple, and powerful methods for finding approximate results of many fractional physical models which arise in applied sciences. In this approach, we do not need the Lagrange multiplier, correction functional, and stationary conditions or to calculate heavy integrals because the results established are noise free, which overcomes the shortcomings of existing methods. It is remarkable that the projected approaches are well-organized analytical methods for finding approximate analytical solutions to complex nonlinear partial differential equations. Finally, we conclude that this scheme, in future, will be taken into account in order to cope with other complex nonlinear fractional-order systems of equations.

Data Availability

The numerical data used to support the findings of this study are included within the article.

Conflicts of Interest

The authors declare that there are no conflicts of interest regarding the publication of this article.

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Research Article

Analysis of Metallic Nanoparticles (Cu, Al₂O_{3,} and SWCNTs) on Magnetohydrodynamics Water-Based Nanofluid through a Porous Medium

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In this communication, the effect of the addition of the copper (Cu), aluminum oxide (Al_2O_3), and single-wall carbon nanotubes (SWCNTs) metallic nanoparticles on the magnetohydrodynamics (MHD) water-based flow over a porous elastic surface is explored. The objective of the work is to include the radiative effect that interacts with the metallic nanoparticles due to permeability of the surface. The significance of this study stems from the fact that the design of various equipment, such as nuclear power plants, gas turbines, propulsion devices for aircraft, and missiles, is dependent on radiative heat transfer. To formulate the mathematical modelling, similarity transformations were used, and nonlinear differential equations were obtained. To solve the formulated nonlinear differential equations, the Runge–Kutta fourth-order numerical scheme is used in conjunction with the shooting technique. The behavior of velocity profile and temperature profile has been discussed in detail and also engineering quantities such as Nusselt and Sherwood number which are calculated. Furthermore, the addition of metallic nanoparticles enhanced the nanofluid properties for energy transfer enrichment and found many applications in various fields of science and technology.

1. Introduction

Choi introduced a new type of fluid called nanofluid in 1995, which has amazing thermal conductivity properties. The goal of the concept is to saturate nanosized particles in conventional fluids known as base fluids. Nanofluids are extremely important in thermal conductivity, heat transfer enhancement, energy, and other thermos-physical properties for industrial applications [1–5]. The heat transfer capacity of a nanofluid after the addition of metallic and nonmetallic nanoparticles in a conventional base fluid was of particular interest to the researchers. Mohebbi et al. [6] investigated the mathematical model of the Heat Transfer Augmentation Associated with Cu/Water Nanofluid in a Channel with Surface Mounted Blocks by using Lattice Boltzmann Method. The numerical method is applied for the forced convection flow and heat transfer of a nanofluid flowing inside a straight circular pipe by Saryazdi et al. [7]. Moreover, Baag and Mishra [8] discussed heat and mass transfer analysis on MHD 3D water-based nanofluid. The prominent examples of nanofluids are ethylene glycols, kerosene, and water. It has been observed that conducting nanofluids presents their special attention because of their use in diversified areas such as biomedical solicitation as tuneable optical filters, drug delivery, and cancer therapy. Watanabe and Pop [9] deliberately presented the magnetohydrodynamic flow of particular fluid for the occurrence of applied magnetic field through a flat plate. Numerical treatment is depicted by Armaghani et al. [10] for the mixed convective flow phenomena of nanofluid within open C-shaped enclosures. For the enhanced properties, they have used CuO nanoparticles which are dispersed within the base fluid water and the enclosure is imposed with constant magnetic field. Furthermore, the influential behavior of the characterizing parameters such as Richardson number and volume concentration affects the flow phenomena as well. The work of Ibrahim and Terbeche [11] leads to bring out the effective properties of the non-Newtonian power-law fluid with due occurrence of the magnetic field. Analytical approach is employed for the solution of the designed problem and numerical methods are useful for the validation of the current result and the convergence criterion.

Fluid flow and heat transfer with non-Newtonian fluids, for example, are a challenge in the modern revolution, particularly in the oil industry, bubble columns and absorption, zymosis, boiling, plastic foam processing [12], etc. However, the possible applications relating to this type of flow can be observed in various industries. The generation of electric power in the corresponding electric power industry is one of the examples that uses the extraction of energy. The governing equations for different non-Newtonian fluid models are amid the utmost complex equations so that the development in mathematical modelling is of great interest nowadays. A time-dependent flow characterized by the several parameters for the nanofluids past an expanding sheet is presented by Andersson et al. [13]. Furthermore, similarity approach for the complex unsteady flow problem past over an expanding sheet is carried out by Elbashbeshy and Bazid [14]. Thermophoresis and Brownian motion effect on the flow of nanofluid through a vertical plate has been studied by Kuznetsov and Nield [15]. They pointed out that the cooling rate of the plate decreases due to decrement in strengthens of thermophoresis and Brownian motion. Heidary and Kermani [16] studied the effect of solid volume fraction of nanofluid and magnetic strength. They examined that existence of magnetic field and nanofluid could significantly enhance heat transfers properties of the flow phenomena. The thermal properties of the base fluids change appreciable after addition of the metallic nanoparticles and calculate the thermos-physical parameters [17]. Masuda et al. [18] reported that, after addition of ultrafine nanoparticles, there is an alteration in the thermal conductivities and viscosities. Mishra et al. [19] recently studied a chemically reactive nano-micropolar fluid with variable heat sink/ source and slip conditions. Shutaywi and Shah [20] proposed a numerical and mathematical model of a nanofluid that includes entropy formation.

The application of electrically conductive fluid currents is encircled in the field of nanocomposite and metallurgy. The flows of several fluids under the action of magnetic field such as MHD generators, oil exploration, energy extraction, and boundary layer control have attracted many researchers. Metallurgical requirements consist of continuous cooling belts or filaments such as hardening, disperse, and sketching processes for copper wires. It has been noticed that the effects of Coriolis force are larger than those of viscosity and inertia forces in the hydro-magnetic equations of motion in

a rotating environment. Several researchers have been investigated on MHD with various kinds of fluid and geometries. For example, Ibrahim and Negera [21] investigated the upper-convected Maxwell nanofluid flow with slip and MHD effects through a stretching sheet and chemical reaction. Abdal et al. [22] examine the thermo-diffusion with magnetized mixed convection unsteady nanofluid flow through stretching/shrinking surface with heat source and thermal radiation. Ghasemi and Hatami [23] described the solar radiation effects on magnetized stagnation point nanofluid flow through a stretching surface. Some important references related to the proposed topic can be found in [24-28] and several therein. Recently, Upreti et al. [29, 30] considered carbon nanotube nanofluids for the behavior of various physical quantities in different geometries. They have projected the effect of drag force with an interaction of Joule heating and nonuniform heat source/sink. Also, binary chemical reaction with the impact of radiative heat on the flow phenomena over an expanding surface is considered. Sabu et al. [31] investigated the enhancement of heat transfer caused by a thermal and space dependent heat source, magnetic field, and nanoparticles propagating over an elastic spinning disk. Mahanthesh et al. [32] investigated Reiner-Rivlin nanofluid flow through a rotating disk with multiple slips and a distinct heat source.

Therefore, the primary goal of this research is to determine the presence of three metallic nanoparticles (Cu, Al₂O₃, and SWCNTs) in an electrically conducting waterbased nanofluid propagating through a porous medium. Thermal radiation is important in industrial applications. As a matter of fact, the study's novelty stems from the incorporation of thermal radiation as well as an additional heat source/sink within a permeable medium. The mathematical modelling was developed using similarity transformations. The nonlinear differential equations are solved using the Runge-Kutta and shooting techniques. When compared to other similar methods used for nonlinear problems, the current numerical method yields promising results [33, 34]. The graphical interpretation of the velocity and temperature profiles have been discussed in detail and expected results show the excellent industrial applications.

2. Problem Formulation

The time-dependent electrically conducting flow of nanofluids through a permeable medium is presented in this article. For the enhanced feature in heat transfer attempt is made to consider SWCNTs in the water-based nanofluid along with Cu and Al_2O_3 nanoparticles. Moreover, the novelty of the study arises for the inclusion of radiative heat transfer with additional external heat source/sink that enriches the energy profile. The flow through porous elastic surface along the *x*-direction and the transverse magnetic field of uniform strength B_0 is proposed along the normal direction of the surface, i.e., *y*-direction, as given in Figure 1. Due to permeability of the surface, the occurrence of suction/injection has its immense use on the flow phenomena. Following Zhang et al. [35], the proposed assumptions lead to design the model with the boundary conditions as



FIGURE 1: Flow configuration.

$$\begin{split} \frac{\partial u}{\partial x} &+ \frac{\partial v}{\partial y} = 0, \\ \rho_{nf} \left(\frac{\partial u}{\partial t} + u \frac{\partial u}{\partial x} + v \frac{\partial u}{\partial y} \right) = -\frac{\partial p}{\partial x} + \mu_{nf} \frac{\partial^2 u}{\partial y^2} \\ &- \left(\frac{\mu_{nf}}{K} + \sigma_{nf} B_0^2 \right) u, \end{split}$$
(1)
$$\left(\rho c_p \right)_{nf} \left(\frac{\partial T}{\partial t} + u \frac{\partial T}{\partial x} + v \frac{\partial T}{\partial y} \right) = k_{nf} \frac{\partial^2 T}{\partial y^2} - \frac{\partial q_r}{\partial y} \\ &+ Q_0 \left(T - T_{\infty} \right), \end{split}$$

with boundary conditions,

$$u(x, 0, t) = 0,$$

$$v(x, 0, t) = v_0(t),$$

$$-k_{nf} \frac{\partial T(x, 0, t)}{\partial t} = q(x),$$

$$u(x, \infty, t) = U(x, t),$$

$$T(x, \infty, t) = T_{\infty}.$$

$$(2)$$

Here, *u* and *v*, are the components of velocities along *x*– and *y*– direction, *T* is the temperature of the nanofluid, *t* is the time taken, *p* is the fluid pressure, v_0 is a constant, and σ_s and σ_f are the electrical conductivity of the base and nanofluid, respectively.

The physical properties relating to nanofluid such as viscosity, specific heat, density, and conductivity are presented as follows [36]:

$$\mu_{nf} = \frac{\mu_f}{(1-\phi)^{2.5}}, \qquad \rho_{nf} = (1-\phi)\rho_f + \phi\rho_s, (\rho c_p)_{nf} = (1-\phi)(\rho c_p)_f + \phi(\rho c_p)_s,$$

$$\sigma_{nf} = (1-\phi)\sigma_f + \phi\sigma_s, \quad \frac{k_{nf}}{k_f} = \frac{k_s + 2k_f - 2\phi(k_f - k_s)}{k_s + 2k_f + \phi(k_f - k_s)},$$
(3)

where ϕ is the particle concentration, μ_f is the dynamics viscosity, ρ_f and ρ_s are the densities, k_f and k_s are the

thermal conductivities, and the subscripts f and s are for the base fluid and the solid nanoparticles.

$$\rho_{nf}\left(\frac{\partial u}{\partial t} + u\frac{\partial u}{\partial x} + v\frac{\partial u}{\partial y}\right) = \frac{\partial U}{\partial t} + U\frac{\partial U}{\partial x} + \mu_{nf}\frac{\partial^2 u}{\partial y^2} - \left(\frac{\mu_{nf}}{K} + \sigma_{nf}B_0^2\right)(u-U),\tag{4}$$

where

$$\frac{\partial p}{\partial x} = \frac{\partial U}{\partial t} + U \frac{\partial U}{\partial x} - \left(\frac{v_{nf}}{K} + \frac{\sigma_{nf}}{\rho_{nf}} B_0^2\right) U,$$
(5)

Here, *K* is the permeability of the given medium and B_0 is an external magnetic field strength.

The stream function as well as variables for the problem is expressed as (see [36])

4

$$u = \frac{\partial \psi}{\partial y},$$

$$v = -\frac{\partial \psi}{\partial x},$$

$$\eta = y \sqrt{\frac{a}{v_f}},$$

$$\psi = \sqrt{av_f} x f(\eta),$$

$$a = \frac{1}{s+bt},$$

$$T = T_{\infty} + \frac{q_0 x}{k_f} \sqrt{\frac{v_f}{a}} \theta(\eta),$$

$$q(x) = q_0 x,$$

$$v_0(t) = v_i(t).$$
(6)

Using the aforesaid functional expressions, the governing equations are presented as

ı

$$f''' + A_1 \left[\left(f + \frac{b\eta}{2} \right) f'' - f'^2 + b \left(f'^2 - 1 \right) + 1 \right]$$
(7)
+ $(MA_2 + Da) (1 - f') = 0,$
$$\frac{1}{A_3 Pr} \left(\frac{k_{nf}}{k_f} + Nr \right) \theta'' + \left(f + \frac{b\eta}{2} \right) \theta' - \left(f' + \frac{b}{2} + \frac{\delta}{A_3} \right) \theta = 0,$$
(8)

where $M = (\sigma_f / \rho_f b) B_0^2$ is the magnetic parameter, Pr = v_f / κ is the Prandtl number, $Nr = 16\sigma_1 T_{\infty}^3 / 3\kappa k_f$ is the thermal radiation parameter, $Da = v_f / aK$ is the Darcy number with boundary condition, b is the unsteadiness parameter, and $\delta = q_0 x / (\rho c_p)_f U$ is the heat source parameter.

$$\begin{cases} f(0) = fw, \\ f'(0) = 0, \\ \theta'(0) = -\frac{k_f}{k_{nf}}, \\ f'(\infty) \longrightarrow 1, \\ \theta(\infty) \longrightarrow 0, \end{cases}$$

$$(9)$$

where

$$A_{1} = (1 - \phi)^{2.5} \left(1 - \phi + \phi \frac{\rho_{s}}{\rho_{f}} \right),$$

$$A_{2} = (1 - \phi)^{2.5} \left(1 - \phi + \phi \frac{\sigma_{s}}{\sigma_{f}} \right),$$

$$A_{3} = 1 - \phi + \phi \frac{(\rho c p)_{s}}{(\rho c p)_{f}}.$$
(10)

The physical quantities are as follows: $C_f = \tau_w / \rho_f U^2$ is called as skin friction coefficient and $Nu_x = q_w x / k_f (T_w - w_f) / k_f (T_w - w_f)$ T_{∞}) is called local Nusselt number:

$$C_{f} \operatorname{Re}_{x}^{0.5} = \frac{f''(0)}{(1-\phi)^{2.5}},$$

$$Nu_{x} \operatorname{Re}_{x}^{-0.5} = -\left(\frac{k_{nf}}{k_{f}} + Nr\right) \theta'(0).$$
(11)

3. Numerical Methodology

For solving equations (10)-(12), a multistep integration method, i.e., the Runge-Kutta method, with shooting technique has been deployed. In this process, equations (7) and (8) are reduced to a set of ordinary differential equations as defined below:

$$f = y_{1}, f' = y_{2}, f'' = y_{3}, f'''$$

$$= -A_{1} \left[\left(y_{1} + \frac{b\eta}{2} \right) y_{3} - y_{2}^{2} + b \left(y_{1}^{2} - 1 \right) + 1 \right]$$

$$- \left(MA_{2} + Da \right) (1 - y_{1}), \qquad (12)$$

$$\theta = y_{4}, \theta' = y_{5}, \theta'' = A_{3} \Pr \left(- \left(y_{1} + \frac{b\eta}{2} \right) y_{5} + \left(y_{2} + \frac{b}{2} + \frac{\delta}{A_{3}} \right) y_{4} \right) \left(\frac{k_{nf}}{k_{f}} + Nr \right)^{-1},$$

under the boundary condition,

 $y_1 = fw$, $y_2 = 0,$ $y_5 = -\frac{k_f}{k_{nf}},$ (13) $y_2 \longrightarrow 1$, $y_4 \longrightarrow 0$.

Now, it uses only the initial conditions, i.e., for $\eta = 0$. However, due to the unavailability of initial conditions, the assumed initial conditions y_3 and y_4 are to be determined. Therefore, some initial guesses are incorporated to both these values in order to satisfy the boundary conditions at $\eta \rightarrow \infty$ ($\eta = 4$). These corrections are taken care by a corrective procedure that follows a self-iterative process. This procedure is used to implement a more accurate method, i.e., the RK method, with shooting numerical technique. For the computational purpose, the step size is assumed as h = 0.01. Therefore, the accuracy of computation and the convergence criteria are followed.

4. Results and Discussion

An unsteady two-dimensional flow of metallic water-based nanofluids is considered which past a permeable medium for the action of transverse magnetic field is presented. Interaction of Cu and Al₂O₃ nanoparticles along with SWCNTs in base fluid water is dispersed to prepare nanofluid. Incorporation of radiative heat energy enriches the profile in conjunction to the permeable surface. Numerical technique is used to find the solution of the set of equations for the suitable choice of the pertinent parameters. Table 1 displays all the physical properties of both the particles as well as the base fluid. Table 2 present the validation of the present outcomes for the shear rate considering the case of pure fluid as well as the case of nanofluid with the work of Rizwan et al. [36], and this shows a good corroboration. The graphical illustration shows the significant behavior of these parameters associated with the flow phenomena. Furthermore, the tabular simulated results indicate the rate coefficients, i.e., shear rate and Nusselt number. However, throughout the computation, the following values of the parameters are considered as fixed whereas the variation of particular parameters are presented in the corresponding figures, and these are $\phi = 0.2$, b = 0.1, M = 1, Da = 1, fw = 0.5, and $\delta = 1.$

The role of particle concentration due to its appearance through the thermo-physical properties is a vital part of this investigation. Figure 2 describes the significance of particle concentration on the velocity for the Cu, Al₂O₃, and SWCNT-water-based nanofluids. Several characteristics of the suction/injection on each profile are displayed. Here, the parameter fw > 0 represents the role of suction whereas fw < 0 indicates the injection and fw = 0 characterizes the behavior when the flow through impermeable region. The decelerating nature of the profiles shows the increasing width of the bounding surface thickness for the increasing particle concentration. The range of the concentration is treated within $\phi = [0.0, 0.2]$. The impermeability region for the pure fluid is similar to the results obtained by Mishra et al. [19], and it can be obtained by considering fw = 0 and $\phi = 0$. Furthermore, the increasing suction enriches the profiles, but the thickness of the bounding surface decreases; however, injection reveals opposite impact on the profile. It reveals the density of the Cu particles and diminishes the profile width in comparison to the particles of Al₂O₃ and SWCNTs. Figure 3 illustrates the behavior of the

TABLE 1: Thermos-physical properties of base fluid and nanoparticles.

	$\rho(\text{kg/m}^3)$	$c_p(J/\text{kg K})$	k(W/mK)
Pure water	997.1	4179	0.613
Copper (Cu)	8933	385	401
Aluminum oxide (Al ₂ O ₃)	3970	765	40
SWCNTs	2600	425	6600

TABLE 2: Validation of shear rate.

Nanofluida	4	f''(0)	f''(0)	
Nationulus	φ	Rizwan et al. [36]	Present	
	0.0	1.48113419	1.481021	
Cu-H ₂ O	0.1	1.71105504	1.711003	
	0.2	1.75138728	1.751128	
	0.0	1.48113419	1.481021	
Al ₂ O ₃ -H ₂ O	0.1	1.43438455	1.431625	
	0.2	1.33096758	1.330727	
	0.0	1.48113419	1.481021	
SWCNT-H ₂ O	0.1	1.45088235	1.450122	
	0.2	1.35571879	1.355526	

unsteadiness parameter in association with the suction/injection on the nanofluid velocity profiles. Here, $b \neq 0$ indicates the unsteady case on the velocity of the three different water-based nanofluid. An augmentation in the profiles is rendered for the increasing unsteadiness that causes a deceleration in the bounding surface thickness. The profiles of SWCNT nanofluid are lesser than the other nanoparticles of Al₂O₃ and Cu, respectively. Also, for each of the profiles, interestingly, the thickness decelerates more in case of injection in comparison to impermeable and the case of suction successively. Figure 4 portrays the role of magnetic parameter on the nanofluid velocity profiles with the interaction suction/injection. The magnetic field expresses the influence of moving electric charges along with electric current and magnetic materials. In modern technology, there are various applications of magnetic field such as in both the electric motors and generators; the use of rotating magnetic field is important. The profile augments lead to decelerate the thickness of the velocity-bounding surface for the augmented magnetic parameter. This is due to the resistance offered by the resistive force produced with the interaction of magnetic parameter, i.e., the Lorentz force. It is seen that SWCNTs have greater retardation than that of Al₂O₃ and Cu-water nanofluid. However, suction also favors to decelerate the profile significantly than that of injection. Furthermore, Figure 5 examines the significance of the permeability parameter on the nanofluid velocity distribution. Similar to the magnetic parameter, resistive force offered by porosity also causes a similar behavior on the velocity profiles for each of the nanofluids. The influence of the suction/injection has the same tendency on the profiles as described in the earlier description. The control of suction/injection due to the permeable surface is shown in Figure 6 for the velocity distribution of nanofluids. Generally, the pressure differential occurs by the elimination of air from the space. Therefore, the limited pressure is exerted



FIGURE 2: Variation of ϕ on velocity profile.



FIGURE 3: Variation of *b* on velocity profile.

by the external air. The pressure in one part of the system is reduced in comparison to another; there will be force exerts from the fluid of higher pressure region to lower. However, with escalating suction, the pressure increases, and this leads to decelerate the surface thickness, whereas impact is reversed for the case of injection. The case of impermeability is a particular case which validates with the earlier result. The significant characteristics of the controlling parameters on the fluid temperature is observed and presented. The role of these parameters enhances the thermos-physical properties significantly. Therefore, the current study discloses the properties of particle concentration, magnetic and porosity parameters, suction/injection, and unsteadiness parameter. Figure 7 displays the role of particle concentration on the nanofluid temperature with an interaction of suction/injection. The three-layer variation explains the distribution of different parameters on the Cu, Al₂O₃, and SWCNT-water-based nanofluids, respectively. Furthermore, the fluid temperature boosts its maximum trend in case of Cu-water nanofluid since it is well known that Cu is a good conductor



FIGURE 4: Variation of M on velocity profile.



FIGURE 5: Variation of Da on velocity profile.

of heat. Furthermore, with the increase in volume fraction, the profile decelerates significantly. Moreover, suction produces more energy to boost the profile rather than the impermeability of the surface and the case of injection. Figure 8 demonstrates the role of unsteadiness parameter that has important characteristics on the nanofluid temperature. Again, increasing unsteadiness, the fluid temperature decelerates in an order of preference such as Cu, Al_2O_3 , and SWCNT-water nanofluid. Therefore, it suggests that effectiveness of the Cu nanoparticle is higher than that of other nanoparticles presented in this study. This gives a suggestive measure for the increasing thermal properties of the Cu-water nanofluid since the proposed thermal conductivity of the nanofluid enhances due to increase in particle concentration. Figure 9 exhibits the effects of the heat source on the nanofluid temperature distribution for different suctions/injections. The inclusion of additional heat suppresses the fluid temperature. In a comparative analysis, it is marked that the Cu-water nanofluid exhibits its maximum strength than other nanofluids. However, no



FIGURE 6: Variation of fw on velocity profile.



FIGURE 7: Variation of ϕ on temperature profile.

significant change is marked for the variation of suction/ injection. Figure 10 depicts the behavior of the thermal radiation in conjunction to the other contributing parameters on the temperature distributions of the nanofluids. Thermal radiation is due to the release of the electromagnetic waves from the fluid particles that is nothing but the renovation of thermal energy into the electromagnetic energy. With an increase in thermal radiation, the profile rises up and therefore the fluid temperature boosts up. This is because most of the solids and fluids are considered to be the surface phenomena and the interior molecules help to emit the radiations.

Finally, the simulated results of the rate coefficients for several contributing parameters are obtained and presented in Table 3. The nanoparticle concentration enhances the shear rate coefficients whereas heat transfer rate decreases in magnitude. From the tabular results, it is quite clear to see that the rate coefficients are much higher in case of Cu-water nanofluid in comparison to other nanofluids. Furthermore, the resistive forces such as magnetic and porosity of the



FIGURE 8: Variation of b on temperature profile.



FIGURE 9: Variation of δ on temperature profile.



FIGURE 10: Variation of Nr on temperature profile.

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TABLE 5: Numerical	computations	or pr	iysical	quantities	usea in	engineering	applications.

4								Cu-water		Al ₂ O ₃ -water		SWCNTs-water	
φ	M	Da	b	fw	Pr	δ	Nr	C_f	Nu_x	C_f	Nu_x	C_f	Nu_x
0.1	1	0.1	0.5	0.5	6.2	1	0.5	2.6073	-4.3891	2.2782	-4.3874	2.1858	-4.2266
0.15								3.0849	-3.7691	2.5915	-3.7728	2.4523	-3.5700
0.2								3.6065	-3.2190	2.9466	-3.2312	2.7597	-3.0029
0.1	2							2.9684	-4.4195	2.6735	-4.4239	2.5920	-4.2637
	3							3.2863	-4.4443	3.0137	-4.4528	2.9387	-4.2927
	1	0.2						2.6450	-4.3924	2.3200	-4.3914	2.2290	-4.2308
		0.3						2.6820	-4.3956	2.3610	-4.3953	2.2713	-4.2348
		0.1	0.1					2.8167	-4.1521	2.4481	-4.1439	2.3432	-3.9932
			0.3					2.7139	-4.2726	2.3645	-4.2678	2.2657	-4.1119
			0.5	-0.5				1.5958	-2.1490	1.5570	-2.1602	1.5442	-2.1508
				0				2.0563	-3.1186	1.8913	-3.1214	1.8433	-3.0520
				0.5	1			2.6073	-1.5644	2.2782	-1.5587	2.1858	-1.5107
					2			2.6073	-2.2670	2.2782	-2.2602	2.1858	-2.1872
					6.2	2		2.6073	-5.0817	2.2782	-5.0971	2.1858	-4.9396
						3		2.6073	-5.6848	2.2782	-5.7129	2.1858	-5.5553
						1	0.1	2.6073	-4.0441	2.2782	-4.0408	2.1858	-3.8926
							0.3	2.6073	-4.2204	2.2782	-4.2180	2.1858	-4.0634

medium favors to enhance the shear rate and opposite trend is rendered for the heat transfer rate. An increase in suction enriches the rate coefficients significantly.

5. Conclusion

The radiative heat transport phenomenon on the two-dimensional flow of water-based nanofluids over an elastic surface is carried out in the current investigation. Here, the electrically conducting nanofluid past a porous surface embedding with porous matrix is presented. The effect of heat source is also included to examine the heat transfer properties. Numerical approach is employed for the solution of the flow phenomena designed by the proposed model. Furthermore, the important characteristics of the physical parameters are laid down here:

- (i) Comparative analysis shows a pathway for the further investigation of the current problem under study for the behavior of several nanoparticles in the water-based fluid with the interaction of various characterizing parameters.
- (ii) Particle concentration decelerates the velocity distributions causing a special effect to enhance the bounding surface thickness whereas the thermal bounding surface behaves in the reverse order, and

it clarifies that Cu nanoparticle has a greater role in both the profile in comparison to Al_2O_3 and SWCNT nanoparticles.

- (iii) The unsteadiness overshoots the velocity profiles for which the thickness of the bounding surface thickness retards; moreover, similar trend is marked for the temperature distribution. However, steady state conditions preserve maximum magnitude for both the profiles.
- (iv) An augmentation in suction enriches the profiles of velocity in comparison to injection, whereas heat source diminishes the fluid temperature significantly.
- (v) The shear rate coefficient rises with increase in particle concentration, whereas heat transfer rate shows its opposite impact.

Data Availability

No data were used to support this study.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

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Research Article

Suppressing Chaos for a Fractional-Order Chaotic Chemical Reaction Model via PD^{ζ} Controller

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In this work, based on the earlier publications, we build a new fractional-order chemical reaction model. Computer simulations manifest that the fractional-order chemical reaction model presents chaotic behavior under a certain parameter condition. To eliminate the chaotic dynamical property, a suitable fractional-order PD^{ζ} controller with time delay is designed. Regarding the time delay as a bifurcation parameter, we set up a novel delay-independent stability and bifurcation criterion guaranteeing the stability and the creation of Hopf bifurcation of the controlled fractional-order chemical reaction model. The influence of time delay on the stability and Hopf bifurcation of the controlled fractional-order chemical reaction model is revealed. At last, numerical simulations are performed to sustain the rationality of the designed PD^{ζ} controller. The obtained conclusions of this work are completely novel and have immense application prospects in the chaos control of chemical reaction systems. Furthermore, the research idea can also be utilized to suppress the chaos of a lot of fractional-order chaotic models.

1. Introduction

Chaos exists widely in many areas such as climate, physics, chemistry, engineering, economy and finance, complex networks, and population systems [1–6]. The chaotic phenomenon depends on the initial condition of the original system. The chaotic behavior occurring in the nonlinear dynamical systems owns the very complex property and unpredictability. In many cases, chaotic behavior is not what we want in our practical life. Thus, a natural problem arises: how to suppress the chaotic behavior of the original system has been an important theme in many disciplines. During the past several decades, suppressing chaos has received great attention from many scholars. For example, Chen [7] controlled the chaos via a simple adaptive feedback control technique. Du et al. [8] applied the phase space compression method to control the chaos of an economic model. In 1990, Yorke [9] utilized Ott, Grebogi, and Yorke (OGY) control approach to control the chaos of a chaotic model. Paula and Savi [10] designed an extended time-delayed feedback control approach to suppress the chaos of a nonlinear pendulum. For more detailed literature on this aspect, one can refer [11–14].

In 1996, Geysermans and Baras [15] proposed a homogeneous chaotic Wilamowski–Rossler model. The balance equations of this model have a well-defined microscopic counterpart and all the reaction follows the following "elementary" steps:

$$\begin{cases} \mathscr{A}_{1} + \mathscr{X} \xrightarrow{\kappa_{1}} 2\mathscr{X}, \mathscr{A}_{1} + \mathscr{X} \xleftarrow{\kappa_{-1}} 2\mathscr{X}, \\ \mathscr{X} + \mathscr{Y} \xrightarrow{\kappa_{2}} 2\mathscr{Y}, \\ \mathscr{A}_{5} + \mathscr{Y} \xrightarrow{\kappa_{3}} \mathscr{A}_{2}, \\ \mathscr{X} + \mathscr{X} \xrightarrow{\kappa_{4}} \mathscr{A}_{3}, \\ \mathscr{A}_{4} + \mathscr{Z} \xrightarrow{\kappa_{5}} 2\mathscr{Z}, \mathscr{A}_{4} + \mathscr{Z} \xleftarrow{\kappa_{-5}} 2\mathscr{Z}. \end{cases}$$
(1)

System (1) includes two autocatalytic steps involving constituents \mathscr{X} and \mathscr{X} , coupled via three other steps involving the three constituents \mathscr{X} , \mathscr{X} , and \mathscr{Y} . The initial $(\mathscr{A}_1, \mathscr{A}_4, \mathscr{A}_5)$ and final $(\mathscr{A}_2, \mathscr{A}_3)$ product concentrations remain fixed. The distance from thermodynamic equilibrium is controlled by the values of $\mathscr{A}_1, \mathscr{A}_2, \mathscr{A}_3, \mathscr{A}_4, \mathscr{A}_5$. $\kappa_{\pm i}$ (i = 1, 2, 3, 4, 5) stands for the rate constant. In model (1), there are 15 free parameters. To reduce the number of free

parameters, Geysermans and Baras [15] selected the rate coefficients $\kappa_{-2} = 0$, $\kappa_{-3} = 0$, and $\kappa_{-4} = 0$. Note that the last two relations imply that \mathscr{A}_2 and \mathscr{A}_3 are continuously removed from the reactor [15, 16].

Assuming that there exists an ideal mixture and a wellstirred reactor, then the macroscopic rate equations of model (1) can be expressed as follows:

$$\begin{cases} \frac{\mathrm{d}u_{1}(t)}{\mathrm{d}t} = \beta_{1}u_{1}(t) - \kappa_{-1}u_{1}^{2}(t) - u_{1}(t)u_{2}(t) - u_{1}(t)u_{3}(t),\\\\ \frac{\mathrm{d}u_{2}(t)}{\mathrm{d}t} = u_{1}(t)u_{2}(t) - \beta_{5}u_{2}(t),\\\\ \frac{\mathrm{d}u_{3}(t)}{\mathrm{d}t} = \beta_{4}u_{3}(t) - u_{1}(t)u_{3}(t) - k_{-5}u_{3}^{2}(t), \end{cases}$$

$$(2)$$

where $u_1(t)$, $u_2(t)$, and $u_3(t)$ stand for the mole fractions of \mathcal{X} , \mathcal{Y} , and \mathcal{Z} at the time *t*. The rate constants κ_1 , κ_3 , and κ_5 are incorporated in the parameters β_1 , β_5 , and β_4 (e.g., $\beta_1 = \kappa_1[\mathcal{A}_1], \ldots, o$) and $\beta_1 > 0$, $\beta_4 > 0$, $\beta_5 > 0$, $\kappa_{-1} > 0$, $\kappa_{-5} > 0$ stand for the constants. In detail, one can refer [15, 16]. In 2015, Xu and Wu [17] dealt with the bifurcation control of chaos for model (2) via three-time delay feedback controllers. Namely, they considered the following controller chemical model:

$$\begin{cases} \frac{du_{1}(t)}{dt} = \beta_{1}u_{1}(t) - \kappa_{-1}u_{1}^{2}(t) - u_{1}(t)u_{2}(t) - u_{1}(t)u_{3}(t) + \mu_{1}[u_{1}(t) - u_{1}(t-\theta)], \\ \frac{du_{2}(t)}{dt} = u_{1}(t)u_{2}(t) - \beta_{5}u_{2}(t) + \mu_{2}[u_{2}(t) - u_{2}(t-\theta)], \\ \frac{du_{3}(t)}{dt} = \beta_{4}u_{3}(t) - u_{1}(t)u_{3}(t) - k_{-5}u_{3}^{2}(t) + \mu_{3}[u_{3}(t) - u_{3}(t-\theta)], \end{cases}$$
(3)

where μ_i (*i* = 1, 2, 3) stands for a real constant and θ is a delay.

It is worth mentioning that all the literature above (see [15–17]) are only concerned with the integer-order chemical models and they are not concerned with the fractional-order chemical models. Recent studies have shown that fractionalorder differential equation is deemed as a more effective tool to portray natural phenomena than the classical integerorder ones since it has great advantages in memory trait and hereditary property of numerous materials and development processes [18-20]. Recently, fractional-order dynamical systems have displayed great application in lots of areas such as network systems, intelligent control, physical science, biological engineering, chemistry, finance, and so on [21-26]. Rich achievements on fractional-order dynamical models have been obtained. For example, Ke [27] dealt with the Mittag–Leffler stability and asymptotic ω -periodicity for a class of fractional-order inertial delayed neural networks. Du and Lu [28] focused on the finite-time stability for fractional-order fuzzy delayed cellular neural networks. Xiao et al. [29] probed into the bifurcation control problem for fractional-order small-world networks; Huang et al. [30] studied the bifurcation problem of fractional-order delayed neural networks. In detail, we refer the readers to [31-34].

Inspired by the exploration above and on the basis of system (2), in order to describe the continuous change

process of the mole fractions of \mathcal{X}, \mathcal{Y} , and \mathcal{Z} and characterize the memory trait and hereditary property of the variables \mathcal{X}, \mathcal{Y} , and \mathcal{Z} , we modify system (2) as the following fractional-order form:

$$\begin{bmatrix} \frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}} = \beta_{1}u_{1}(t) - \kappa_{-1}u_{1}^{2}(t) - u_{1}(t)u_{2}(t) - u_{1}(t)u_{3}(t), \\ \frac{d^{\zeta} u_{2}(t)}{dt^{\zeta}} = u_{1}(t)u_{2}(t) - \beta_{5}u_{2}(t), \\ \frac{d^{\zeta} u_{3}(t)}{dt^{\zeta}} = \beta_{4}u_{3}(t) - u_{1}(t)u_{3}(t) - k_{-5}u_{3}^{2}(t), \quad (4)$$

where $\zeta \in (0, 1]$. The research indicates that when $\zeta = 0.97, \beta_1 = 30, \kappa_{-1} = 0.55, \beta_5 = 9.5, \beta_4 = 16.5, \kappa_{-5} = 0.5$, then there is a chaotic phenomenon in system (3). The software simulation figures are presented in Figure 1.

In this current work, we are to deal with the chaos control of system (4) by virtue of a fractional-order PD^{ζ} controller. The key contributions of this study are as follows:

(i) On the basis of the earlier studies, a new fractionalorder chaotic chemical reaction model is set up



FIGURE 1: Software simulation figures of system (4) with $\zeta = 0.97$, $\beta_1 = 30$, $\kappa_{-1} = 0.55$, $\beta_5 = 9.5$, $\beta_4 = 16.5$, $\kappa_{-5} = 0.5$.

- (ii) The chaotic phenomenon of system (4) is suppressed by means of an appropriate fractional-order PD^ζ controller
- (iii) The study approach can be utilized to suppress the chaos of lots of fractional-order dynamical models in many subjects

This manuscript can be arranged as follows: some prerequisite theory on the fractional-order differential equation is prepared in Section 2; in Section 3, we prove the existence and uniqueness of the solution of system (4); in Section 4, the chaos of system (4) is suppressed via fractional-order PD^{ζ} controller and a delay-independent sufficient condition that ensures the stability and the creation of Hopf bifurcation of the fractional-order controlled chaotic chemical reaction model is built; in Section 5, software simulation results are presented to sustain the established conclusions; and Section 6 completes this article.

2. Preliminary Knowledge

In this part, we present some indispensable theories on a fractional-order differential equation.

Definition 1 (see [35]). The fractional type integral of the order ζ of the function $u(\xi)$ is given by

$$\mathcal{J}^{\zeta}u(\xi) = \frac{1}{\Gamma(\zeta)} \int_{\xi_0}^{\xi} (\xi - \nu)^{\zeta - 1} u(\nu) \mathrm{d}\nu, \tag{5}$$

where $\xi > \xi_0, \zeta > 0$ and $\Gamma(\nu) = \int_0^\infty s^{\nu-1} e^{-s} ds$.

Definition 2 (see [35]). The Caputo fractional order derivative of the order ζ of the function $u(v) \in ([v_0, \infty), R)$ is defined as follows:

$$\mathscr{D}^{\zeta}u(\nu) = \frac{1}{\Gamma(\kappa - \zeta)} \int_{\nu_0}^{\nu} \frac{u^{(\kappa)}(s)}{(\nu - s)^{\zeta - \kappa + 1}} \mathrm{d}s, \tag{6}$$

where $\nu \ge \nu_0$ and κ represents a positive integer $(\zeta \in [\kappa - 1, \kappa))$. In particular, if $\zeta \in (0, 1)$, then

$$\mathscr{D}^{\zeta}u(\nu) = \frac{1}{\Gamma(1-\zeta)} \int_{\nu_0}^{\nu} \frac{u'(s)}{(\nu-s)^{\zeta}} \mathrm{d}s.$$
(7)

Lemma 1 (see [36]). Consider the fractional-order system $\mathfrak{D}^{\zeta} w = \mathscr{F} w, w(0) = w_0$ where $\zeta \in (0, 1), w \in \mathbb{R}^l, \mathscr{F} \in \mathbb{R}^{|x|}$. Assuming that $\chi_i (i = 1, 2, ..., l)$ is the root of the characteristic equation of $\mathfrak{D}^{\zeta} w = \mathscr{F} w$, then the equilibrium point of the system $\mathfrak{D}^{\zeta} w = \mathscr{F} w$ is locally asymptotically stable if $|\arg(\chi_i)| > (\zeta \pi/2) (i = 1, 2, ..., l)$ and the equilibrium point of the system $\mathfrak{D}^{\zeta} w = \mathscr{F} w$ is stable if $|\arg(\chi_i)| > (\zeta \pi/2) (i = 1, 2, ..., l)$ and all critical eigenvalues that satisfy $|\arg(\chi_i)| = (\zeta \pi/2) (i = 1, 2, ..., l)$ own geometric multiplicity one.

3. Existence and Uniqueness of the Solution of System (4)

In this section, we will prove the existence and uniqueness of the solution of system (4).

Theorem 1. Let $\Lambda = \{(u_1, u_2, u_3) \in \mathbb{R}^3 : \max\{|u_1|, |u_2|, |u_3|\} \le A\}$, where A > 0 is a constant. $\forall (u_{10}, u_{20}, u_{30}) \in \Lambda$, system (4) with the initial value (u_{10}, u_{20}, u_{30}) has a unique solution $U = (u_1, u_2, u_3) \in \Lambda$.

Proof. Define the following mapping:

$$f(U) = (f_1(U), f_2(U), f_3(U)),$$
(8)

where

$$\begin{cases} f_1(U) = \beta_1 u_1(t) - \kappa_{-1} u_1^2(t) - u_1(t) u_2(t) - u_1(t) u_3(t), \\ f_2(U) = u_1(t) u_2(t) - \beta_5 u_2(t), \\ f_3(U) = \beta_4 u_3(t) - u_1(t) u_3(t) - k_{-5} u_3^2(t). \end{cases}$$
(9)

 $\forall U, \overline{U} \in \Lambda$, one obtains

$$\begin{split} \|f(U) - f(\widetilde{U})\| \\ &= |\beta_{1}u_{1}(t) - \kappa_{-1}u_{1}^{2}(t) - u_{1}(t)u_{2}(t) - u_{1}(t)\overline{u}_{3}(t) \\ &- \left[\beta_{1}\overline{u}_{1}(t) - \kappa_{-1}\overline{u}_{1}^{2}(t) - \overline{u}_{1}(t)\overline{u}_{2}(t) - \overline{u}_{1}(t)\overline{u}_{3}(t)\right]| \\ &+ |u_{1}(t)u_{2}(t) - \beta_{5}u_{2}(t) - \left[\overline{u}_{1}(t)\overline{u}_{2}(t) - \beta_{5}\overline{u}_{2}(t)\right]| \\ &+ |\beta_{4}u_{3}(t) - u_{1}(t)u_{3}(t) - k_{-5}u_{3}^{2}(t) \\ &- \left[\beta_{4}\overline{u}_{3}(t) - \overline{u}_{1}(t)\overline{u}_{3}(t) - k_{-5}\overline{u}_{3}^{2}(t)\right]| \\ \leq (\beta_{1} + 2\kappa_{-1}A + A)|u_{1}(t) - \overline{u}_{1}(t)| + A|u_{2}(t) - \overline{u}_{2}(t)| + A|u_{3}(t) - \overline{u}_{3}(t)| + A|u_{1}(t) - \overline{u}_{1}(t)| \\ &+ (A + \beta_{5})|u_{2}(t) - \overline{u}_{2}(t)| + A|u_{1}(t) - \overline{u}_{1}(t)| \\ &+ (\beta_{4} + A + 2\kappa_{-5}A)|u_{3}(t) - \overline{u}_{3}(t)| \\ = A_{1}|u_{1}(t) - \overline{u}_{1}(t)| + A_{2}|u_{2}(t) - \overline{u}_{2}(t)| + A_{3}|u_{3}(t) - \overline{u}_{3}(t)| \\ \leq A_{0}\|U - \overline{U}\|, \end{split}$$

where

$$\begin{cases}
A_1 = \beta_1 + 2\kappa_{-1}A + 3A, \\
A_2 = 2A + \beta_5, \\
A_3 = \beta_4 + 2A + 2\kappa_{-5}A,
\end{cases}$$
(11)

$$A_0 = \max\{A_1, A_2, A_3\}.$$
 (12)

Then f(U) satisfies Lipschitz condition with respect to U (see [39, 40]). According to Banach fixed point theorem, we know that Theorem 1 is true.

4. Suppressing Chaos via Fractional-Order PD^ζ Controller

In this part, we are to apply a suitable controller to eliminate the chaotic phenomenon of system (4). By virtue of the idea of Tang et al. [37], the fractional-order PD^{ζ} controller can be designed as follows:

$$\psi(t) = \varrho_p u_1(t-\theta) + \varrho_d \frac{\mathrm{d}^{\zeta} u_1(t)}{\mathrm{d}t^{\zeta}},\tag{13}$$

where ϱ_p and $\varrho_d \neq 1$ present the proportional control parameter and the derivative control parameter, respectively, and θ denotes a delay. Adding (13) to the first equation of system (4), we get

$$\begin{cases} \frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}} = \beta_{1} u_{1}(t) - \kappa_{-1} u_{1}^{2}(t) - u_{1}(t) u_{2}(t) - u_{1}(t) u_{3}(t) + \psi(t), \\ \frac{d^{\zeta} u_{2}(t)}{dt^{\zeta}} = u_{1}(t) u_{2}(t) - \beta_{5} u_{2}(t), \\ \frac{d^{\zeta} u_{3}(t)}{dt^{\zeta}} = \beta_{4} u_{3}(t) - u_{1}(t) u_{3}(t) - k_{-5} u_{3}^{2}(t). \end{cases}$$

$$(14)$$

That is

$$\frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}} = \beta_{1} u_{1}(t) - \kappa_{-1} u_{1}^{2}(t) - u_{1}(t) u_{2}(t) - u_{1}(t) u_{3}(t)
+ \varrho_{p} u_{1}(t-\theta) + \varrho_{d} \frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}},
\frac{d^{\zeta} u_{2}(t)}{dt^{\zeta}} = u_{1}(t) u_{2}(t) - \beta_{5} u_{2}(t),
\frac{d^{\zeta} u_{3}(t)}{dt^{\zeta}} = \beta_{4} u_{3}(t) - u_{1}(t) u_{3}(t) - k_{-5} u_{3}^{2}(t).$$
(15)

System (15) can be rewritten as the following form:

$$\frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}} = \frac{\beta_{1}}{1 - \varrho_{d}} u_{1}(t) - \frac{\kappa_{-1}}{1 - \varrho_{d}} u_{1}^{2}(t) - \frac{1}{1 - \varrho_{d}} u_{1}(t) u_{2}(t)
- \frac{1}{1 - \varrho_{d}} u_{1}(t) u_{3}(t) + \frac{\varrho_{p}}{1 - \varrho_{d}} u_{1}(t - \theta),
\frac{d^{\zeta} u_{2}(t)}{dt^{\zeta}} = u_{1}(t) u_{2}(t) - \beta_{5} u_{2}(t),
\frac{d^{\zeta} u_{3}(t)}{dt^{\zeta}} = \beta_{4} u_{3}(t) - u_{1}(t) u_{3}(t) - k_{-5} u_{3}^{2}(t).$$
(16)

It is not difficult to obtain that if the following condition

$$(\mathcal{Q}_1)\beta_4 > \beta_5, (\beta_1 - \kappa_{-1})\kappa_{-5} > \kappa_4 - \kappa_5, \tag{17}$$

holds, then system (16) owns the following unique positive equilibrium $\mathcal{U}(u_1^*, u_2^*, u_3^*)$, where

$$\begin{cases}
 u_1^* = \beta_5, \\
 u_2^* = \frac{(\beta_1 - \kappa_{-1}\beta_5)\kappa_{-5} - \beta_4 + \beta_5}{\kappa_{-5}}, \\
 u_3^* = \frac{\beta_4 - \beta_5}{\kappa_{-5}}.
 \end{cases}$$
(18)

The linear system of (16) around the positive equilibrium $\mathcal{U}(u_1^*, u_2^*, u_3^*)$ takes the following form:

$$\begin{cases} \frac{d^{\zeta} u_{1}(t)}{dt^{\zeta}} = a_{1}u_{1}(t) + a_{2}u_{2}(t) + a_{2}u_{3}(t) + a_{3}u_{1}(t-\theta), \\\\ \frac{d^{\zeta} u_{2}(t)}{dt^{\zeta}} = a_{4}u_{1}(t) + a_{5}u_{2}(t), \\\\ \frac{d^{\zeta} u_{3}(t)}{dt^{\zeta}} = a_{6}u_{1}(t) + a_{7}u_{3}(t), \end{cases}$$
(19)

where

$$\begin{cases} a_{1} = \frac{\beta_{1} - 2\kappa_{-1}u_{1}^{*} - u_{2}^{*} - u_{3}^{*}}{1 - \varrho_{d}}, \\ a_{2} = -\frac{u_{1}^{*}}{1 - \varrho_{d}}, \\ a_{3} = \frac{\varrho_{p}}{1 - \varrho_{d}}, \\ a_{4} = u_{2}^{*}, \\ a_{5} = u_{1}^{*} - \beta_{5}, \\ a_{6} = -u_{3}^{*}, \\ a_{7} = \beta_{4} - u_{1}^{*} - 2\kappa_{5}u_{3}^{*}. \end{cases}$$

$$(20)$$

The characteristic equation of system (19) takes the form

$$\det \begin{bmatrix} s^{\zeta} - a_1 - a_3 e^{-s\theta} & -a_2 & -a_2 \\ -a_4 & s^{\zeta} - a_5 & 0 \\ -a_6 & 0 & s^{\zeta} - a_7 \end{bmatrix} = 0.$$
(21)

Then,

$$s^{3\zeta} + b_1 s^{2\zeta} + b_2 s^{\zeta} + b_3 + (c_1 s^{2\zeta} + c_2 s^{\zeta} + c_3) e^{-s\theta} = 0, \quad (22)$$

where

$$\begin{cases} b_1 = -(a_1 + a_5 + a_7), \\ b_2 = a_5 a_7 + a_1 a_5 + a_1 a_7 - a_2 a_6 - a_2 a_4, \\ b_3 = a_2 a_5 a_6 + a_2 a_4 a_7 - a_1 a_5 a_7, \\ c_1 = -a_3, \\ c_2 = a_3 (a_5 + a_7), \\ c_3 = -a_3 a_5 a_7. \end{cases}$$
(23)

When $\theta = 0$, then (22) becomes

$$\lambda^{3} + (b_{1} + c_{1})\lambda^{2} + (b_{2} + c_{2})\lambda + b_{3} + c_{3} = 0.$$
 (24)

Assuming that

$$(\mathcal{Q}_2) \begin{cases} b_1 + c_1 > 0, \\ (b_1 + c_1)(b_2 + c_2) > b_3 + c_3, \\ (b_3 + c_3)[(b_1 + c_1)(b_2 + c_2) - (b_3 + c_3)] > 0, \end{cases}$$
 (25)

is true, then the three roots $\lambda_1, \lambda_2, \lambda_3$ of (24) satisfy $|\arg(\lambda_1)| > (\zeta \pi/2), |\arg(\lambda_2)| > (\zeta \pi/2)$, and $|\arg(\lambda_3)| > (\zeta \pi/2)$. By virtue of Lemma 1, we can conclude that the positive equilibrium point $\mathcal{U}(u_1^*, u_2^*, u_3^*)$ of system (14) is locally asymptotically stable when $\theta = 0$.

Assume that $s = i\rho = \rho \left(\cos \left(\zeta \pi / 2 \right) + i \sin \left(\pi / 2 \right) \right)$ is the root of equation (22). It follows from (22) that

$$\rho^{3\zeta} \left(\cos \frac{3\zeta \pi}{2} + i \sin \frac{3\zeta \pi}{2} \right) + b_1 \rho^{2\zeta} \left(\cos \zeta \pi + i \sin \zeta \pi \right)$$
$$+ b_2 \rho^{\zeta} \left(\cos \frac{\zeta \pi}{2} + i \sin \frac{\zeta \pi}{2} \right) + b_3$$
$$+ \left[c_1 \rho^{2\zeta} \left(\cos \zeta \pi + i \sin \zeta \pi \right) + c_2 \rho^{\zeta} \left(\cos \frac{\zeta \pi}{2} + i \sin \frac{\zeta \pi}{2} \right) + c_3 \right]$$
$$\times \left(\cos \rho \theta - i \sin \rho \theta \right) = 0.$$

Then,

$$\begin{cases} \mathscr{G}_{1} \cos \rho \theta + \mathscr{G}_{2} \sin \rho \theta = \mathscr{H}_{1}, \\ \mathscr{G}_{2} \cos \rho \theta - \mathscr{G}_{1} \sin \rho \theta = \mathscr{H}_{2}, \end{cases}$$
(27)

where

$$\begin{cases} \mathscr{G}_{1} = d_{1}\rho^{2\zeta} + d_{2}\rho^{\zeta} + d_{3}, \\ \mathscr{G}_{2} = d_{4}\rho^{2\zeta} + d_{5}\rho^{\zeta}, \\ \mathscr{H}_{1} = e_{1}\rho^{3\zeta} + e_{2}\rho^{2\zeta} + e_{3}\rho^{\zeta} + e_{4}, \\ \mathscr{H}_{2} = e_{5}\rho^{3\zeta} + e_{6}\rho^{2\zeta} + e_{7}\rho^{\zeta}, \end{cases}$$
(28)

where

$$\begin{cases} d_{1} = c_{1} \cos \zeta \pi, \\ d_{2} = c_{2} \cos \frac{\zeta \pi}{2}, \\ d_{3} = c_{3}, \\ d_{4} = c_{1} \sin \zeta \pi, \\ d_{5} = c_{2} \sin \frac{\zeta \pi}{2}, \\ e_{1} = -\cos \frac{\zeta \pi}{2}, \\ e_{2} = -b_{1} \cos \zeta \pi, \\ e_{3} = -b_{2} \cos \frac{\zeta \pi}{2}, \\ e_{4} = -b_{3}, \\ e_{5} = -\sin \frac{\zeta \pi}{2}, \\ e_{6} = -b_{1} \sin \zeta \pi, \\ e_{7} = -b_{2} \sin \frac{\zeta \pi}{2}. \end{cases}$$
(29)

It follows from (27) that

$$\cos \rho \theta = \frac{\mathscr{H}_1 \mathscr{G}_1 + \mathscr{H}_2 \mathscr{G}_2}{\mathscr{G}_1^2 + \mathscr{G}_2^2},\tag{30}$$

$$\mathscr{G}_1^2 + \mathscr{G}_2^2 = \mathscr{H}_1^2 + \mathscr{H}_2^2.$$
(31)

By virtue of (28) and (31), one gets

$$\left(d_1 \rho^{2\zeta} + d_2 \rho^{\zeta} + d_3 \right)^2 + \left(d_4 \rho^{2\zeta} + d_5 \rho^{\zeta} \right)^2 =$$

$$\left(e_1 \rho^{3\zeta} + e_2 \rho^{2\zeta} + e_3 \rho^{\zeta} + e_4 \right)^2 + \left(e_5 \rho^{3\zeta} + e_6 \rho^{2\zeta} + e_7 \rho^{\zeta} \right)^2,$$
(32)

which leads to

(26)

$$\varepsilon_1 \rho^{6\zeta} + \varepsilon_2 \rho^{5\zeta} + \varepsilon_3 \rho^{4\zeta} + \varepsilon_4 \rho^{3\zeta} + \varepsilon_5 \rho^{2\zeta} + \varepsilon_6 \rho^{\zeta} + \varepsilon_7 = 0, \quad (33)$$

where

$$\begin{cases} \epsilon_{1} = e_{1}^{2} + e_{5}^{2}, \\ \epsilon_{2} = 2(e_{1}e_{2} + e_{5}e_{6}), \\ \epsilon_{3} = e_{2}^{2} + e_{6}^{2} - d_{1}^{2} - d_{4}^{2} + 2(e_{1}e_{3} + e_{5}e_{7}), \\ \epsilon_{4} = 2(e_{1}e_{4} + e_{2}e_{3} + e_{6}e_{7} - d_{1}d_{2} - d_{4}d_{5}), \\ \epsilon_{5} = e_{3}^{2} + e_{7}^{2} - d_{2}^{2} - d_{5}^{2} + 2(e_{2}e_{4} - d_{1}d_{3}), \\ \epsilon_{6} = 2(e_{3}e_{4} - d_{2}d_{3}), \\ \epsilon_{7} = e_{4}^{2} - d_{3}^{2}. \end{cases}$$
(34)

Set

$$\Theta(\rho) = \epsilon_1 \rho^{6\zeta} + \epsilon_2 \rho^{5\zeta} + \epsilon_3 \rho^{4\zeta} + \epsilon_4 \rho^{3\zeta} + \epsilon_5 \rho^{2\zeta} + \epsilon_6 \rho^{\zeta} + \epsilon_7.$$
(35)

Assuming that

$$(\mathcal{Q}_3) |e_4| < |d_3| \tag{36}$$

is true, since $\lim_{\rho \to \infty} \Theta(\rho) = +\infty$, then equation (33) owns at least one real positive root. So equation (22) has at least

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one pair of pure roots. Making use of Sun et al. [38], one can easily establish the conclusion as follows.

Lemma 2. (a) Supposing that $\epsilon_k > 0$ (k = 1, 2, 3, 4, 5, 6), equation (22) owns no root with zero real parts for $\theta \ge 0$. (b) Supposing that (\mathbb{Q}_3) holds and $\epsilon_k > 0$ (k = 1, 2, 3, 4, 5), then equation (22) owns a pair of purely imaginary roots $\pm i\rho_0$ if $\theta = \theta_0^{(l)}$ (l = 1, 2, ...,) where

$$\theta_0^{(l)} = \frac{1}{\rho_0} \left[\arccos\left(\frac{\mathscr{H}_1 \mathscr{G}_1 + \mathscr{H}_2 \mathscr{G}_2}{\mathscr{G}_1^2 + \mathscr{G}_2^2}\right) + 2l\pi \right], \tag{37}$$

where $l = 0, 1, ..., and \rho_0 > 0$ represents the unique zero of $\Theta(\rho)$.

Denote $\theta_0 = \theta_0^{(0)}$. Now the following hypothesis is given:

$$(\mathcal{Q}_4) \mathcal{C}_{1R} \mathcal{C}_{2R} + \mathcal{C}_{1I} \mathcal{C}_{2I} > 0, \tag{38}$$

where

$$\begin{cases} \mathscr{C}_{1R} = 3\zeta\rho_{0}^{3\zeta^{-1}}\cos\frac{(3\zeta^{-1})\pi}{2} + 2\zeta_{1}\rho_{0}^{2\zeta^{-1}}\cos\frac{(2\zeta^{-1})\pi}{2} \\ + \zeta_{2}\rho_{0}^{\zeta^{-1}}\cos\frac{(\zeta^{-1})\pi}{2} + \left[2\zeta_{1}\rho_{0}^{2\zeta^{-1}}\cos\frac{(2\zeta^{-1})\pi}{2} + \zeta_{2}\rho_{0}^{\zeta^{-1}}\cos\frac{(\zeta^{-1})\pi}{2}\right]\cos\rho_{0}\theta_{0} + \sin\rho_{0}\theta_{0} \\ \times \left[2\zeta_{0}\rho_{0}^{2\zeta^{-1}}\sin\frac{(2\zeta^{-1})\pi}{2} + \zeta_{2}\rho_{0}^{\zeta^{-1}}\sin\frac{(\zeta^{-1})\pi}{2}\right], \\ \mathscr{C}_{1I} = 3\zeta\rho_{0}^{3\zeta^{-1}}\sin\frac{(3\zeta^{-1})\pi}{2} + 2\zeta_{1}\rho_{0}^{2\zeta^{-1}}\sin\frac{(2\zeta^{-1})\pi}{2} \\ + \zeta_{2}\rho_{0}^{\zeta^{-1}}\sin\frac{(\zeta^{-1})\pi}{2} - \left[2\zeta_{0}\rho_{0}^{2\zeta^{-1}}\cos\frac{(2\zeta^{-1})\pi}{2} + \zeta_{2}\rho_{0}^{\zeta^{-1}}\cos\frac{(\zeta^{-1})\pi}{2}\right]\sin\rho_{0}\theta_{0} + \cos\rho_{0}\theta_{0} \\ \times \left[2\zeta_{0}\rho_{0}^{2\zeta^{-1}}\sin\frac{(2\zeta^{-1})\pi}{2} + \zeta_{2}\rho_{0}^{\zeta^{-1}}\sin\frac{(\zeta^{-1})\pi}{2}\right], \\ \mathscr{C}_{2R} = \left(c_{1}\rho_{0}^{2\zeta}\cos\zeta\pi + c_{2}\rho_{0}^{\zeta}\cos\frac{\zeta\pi}{2} + c_{3}\right)\rho_{0}\cos\rho_{0}\theta_{0} \\ - \left(c_{1}\rho_{0}^{2\zeta}\sin\zeta\pi + c_{2}\rho_{0}^{\zeta}\sin\frac{\zeta\pi}{2} + c_{3}\right)\rho_{0}\cos\rho_{0}\theta_{0} \\ + \left(c_{1}\rho_{0}^{2\zeta}\sin\zeta\pi + c_{2}\rho_{0}^{\zeta}\sin\frac{\zeta\pi}{2} + c_{3}\right)\rho_{0}\sin\rho_{0}\theta_{0}. \end{cases}$$

Lemma 3. Let $s(\theta) = \phi_1(\theta) + i\phi_2(\theta)$ be the root of (22) at $\theta = \theta_0$ satisfying $\phi_1(\theta_0) = 0, \phi_2(\theta_0) = \rho_0$, then $Re(ds/d\theta)|_{\theta=\theta_0, \rho=\rho_0} > 0$.

Proof. Making use of (22), we get

$$(3\zeta s^{3\zeta-1} + 2\zeta b_1 s^{2\zeta-1} + \zeta b_2 s^{\zeta-1}) \frac{\mathrm{d}s}{\mathrm{d}\theta} + (2\zeta c_1 s^{2\zeta-1} + \zeta c_2 s^{\zeta-1}) e^{-s\theta} \frac{\mathrm{d}s}{\mathrm{d}\theta} - e^{-s\theta} \left(\frac{\mathrm{d}s}{\mathrm{d}\theta}\theta + s\right) (c_1 s^{2\zeta} + c_2 s^{\zeta} + c_3) = 0,$$

$$(40)$$

$$\left[3\zeta s^{3\zeta-1} + 2\zeta b_1 s^{2\zeta-1} + \zeta b_2 s^{\zeta-1} + \left(2\zeta c_1 s^{2\zeta-1} + \zeta c_2 s^{\zeta-1} \right) e^{-s\theta} - \theta e^{-s\theta} \left(c_1 s^{2\zeta} + c_2 s^{\zeta} + c_3 \right) \right] \frac{\mathrm{d}s}{\mathrm{d}\theta}$$

$$= s e^{-s\theta} \left(c_1 s^{2\zeta} + c_2 s^{\zeta} + c_3 \right).$$

$$(41)$$

which leads to

Then,

$$\left(\frac{\mathrm{d}s}{\mathrm{d}\theta}\right)^{-1} = \frac{\mathscr{C}_1(s)}{\mathscr{C}_2(s)} - \frac{\theta}{s},\tag{42}$$

where

$$\begin{cases} \mathscr{C}_{1}(s) = 3\zeta s^{3\zeta-1} + 2\zeta b_{1} s^{2\zeta-1} + \zeta b_{2} s^{\zeta-1} \\ + (2\zeta c_{1} s^{2\zeta-1} + \zeta c_{2} s^{\zeta-1}) e^{-s\theta}, \\ \mathscr{C}_{2}(s) = s e^{-s\theta} (c_{1} s^{2\zeta} + c_{2} s^{\zeta} + c_{3}). \end{cases}$$
(43)

Then,

$$\operatorname{Re}\left[\left(\frac{\mathrm{d}s}{\mathrm{d}\theta}\right)^{-1}\right]_{\theta=\theta_{0},\rho=\rho_{0}} = \operatorname{Re}\left[\frac{\mathscr{C}_{1}\left(s\right)}{\mathscr{C}_{2}\left(s\right)}\right]_{\theta=\theta_{0},\rho=\rho_{0}}$$

$$= \frac{\mathscr{C}_{1R}\mathscr{C}_{2R} + \mathscr{C}_{1I}\mathscr{C}_{2I}}{\mathscr{C}_{2R}^{2} + \mathscr{C}_{2I}^{2}}.$$

$$(44)$$

In view of (Q_4) , we have

$$\operatorname{Re}\left[\left(\frac{\mathrm{d}s}{\mathrm{d}\theta}\right)^{-1}\right]_{\theta=\theta_{0},\rho=\rho_{0}} > 0, \qquad (45)$$

which completes the proof.

Making use of Lemma 1, we can easily obtain the following conclusion. $\hfill \Box$

Theorem 2. Supposing that $(\mathcal{Q}_1)-(\mathcal{Q}_4)$ hold, then the positive equilibrium point $\mathcal{U}(u_1^*, u_2^*, u_3^*)$ of system (16) is locally

asymptotically stable if the time delay θ lies in the interval $[0, \theta_0)$ and the Hopf bifurcation phenomenon of system (16) will arise near the positive equilibrium point $\mathcal{U}(u_1^*, u_2^*, u_3^*)$ if $\theta = \theta_0$.

Remark 1. Xu and Wu [17] dealt with the chaos control for an integer-order chaotic chemical reaction model by timedelay feedback control technique. This manuscript deals with the chaos control issue for a fractional-order chaotic chemical reaction model via a fractional-order PD^{ζ} controller. The model and the research approach is very different from those in [17]. From this viewpoint, we think that the obtained results and the research method of this manuscript supplement the work of [17] and promote the development of the chaos control theory of fractional-order differential equation to some degree.

Remark 2. In this paper, we use the fractional-order PD^{ζ} controller to control the chaos of the fractional-order chaotic chemical reaction model (4). Compared with the time delay feedback controller, the fractional-order PD^{ζ} controller has more adjustable parameters and then can control the chaos of model (4) neatly.

5. Example

Consider the following controlled fractional-order chaotic chemical reaction model:



FIGURE 2: Computer simulation figures of the controlled fractional-order chaotic chemical reaction model (46) with $\theta = 0.20 < \theta_0 = 0.25$. The blue line represents $u_1(t)$, the red line represents $u_2(t)$, and the green line represents $u_3(t)$.



FIGURE 3: Computer simulation figures of the controlled fractional-order chaotic chemical reaction model (46) with $\theta = 0.28 > \theta_0 = 0.25$. The blue line represents $u_1(t)$, the red line represents $u_2(t)$, and the green line represents $u_3(t)$.



FIGURE 4: Bifurcation plot of the controlled fractional-order chaotic chemical reaction model (46): θ - u_1 .

$$\begin{cases} \frac{d^{\zeta}u_{1}(t)}{dt^{\zeta}} = \beta_{1}u_{1}(t) - \kappa_{-1}u_{1}^{2}(t) - u_{1}(t)u_{2}(t) - u_{1}(t)u_{3}(t) \\ + \varrho_{p}u_{1}(t-\theta) + \varrho_{d}\frac{d^{\zeta}u_{1}(t)}{dt^{\zeta}}, \\ \frac{d^{\zeta}u_{2}(t)}{dt^{\zeta}} = u_{1}(t)u_{2}(t) - \beta_{5}u_{2}(t), \\ \frac{d^{\zeta}u_{3}(t)}{dt^{\zeta}} = \beta_{4}u_{3}(t) - u_{1}(t)u_{3}(t) - k_{-5}u_{3}^{2}(t), \end{cases}$$
(46)

where $\zeta = 0.97, \beta_1 = 30, \kappa_{-1} = 0.55, \beta_5 = 9.5, \beta_4 = 16.5, \kappa_{-5} = 0.5$. Let $\rho_p = 0.5, \rho_d = 0.9$. It is not difficult to obtain the unique positive equilibrium point of system (46) is $\mathcal{U}(9.5,10.775,14)$. By direct computation via MATLAB software, we can easily get $\rho_0 = 4.0239$ and $\theta_0 = 0.25$. The three assumptions $(Q_1) - (Q_4)$ of Theorem 2 are easily verified to be right. So we can conclude that the positive equilibrium point $\mathcal{U}(9.5,$ 10.775,14) of system (46) is locally asymptotically stable if the time delay θ lies in the interval [0,0.25) and the Hopf bifurcation phenomenon for system (46) will arise near the positive equilibrium point $\mathscr{U}(9.5,10.775,14)$ if $\theta=0.25$. In this paper, we use the predictor-correctors approach [39, 41, 42] to discretize system (46) and by virtue of the MATLAB software to carry out numerical simulations. In order to display these results, we select two sets of different delay parameters. Firstly, we choose $\theta = 0.20 < \theta_0 = 0.25$, and the software simulation plots are presented in Figure 2, which implies that $u_1 \longrightarrow 9.5, u_2 \longrightarrow 10.775, u_3 \longrightarrow 14$ as the time *t* tends to infinity. From the chemical point of view, the mole fraction of the constituent $\mathcal X$ will be close to 9.5, the mole fraction of the constituent \mathcal{Y} will be close to 10.775, and the mole fraction of the constituent \mathcal{Z} will be close to 14. Secondly, we choose $\theta = 0.28 > \theta_0 = 0.25$, and the software simulation plots are presented in Figure 3, which implies that a Hopf bifurcation periodic solution of system (46) will arise near the positive equilibrium point $\mathcal{U}(9.5, 10.775, 14)$ as the time *t* tends to infinity. From the chemical point of view, the mole fraction of the constituent \mathcal{X} , the mole fraction of the constituent \mathcal{Y} , and the mole fraction of the constituent $\mathcal Z$ will remain periodically oscillatory situations near the values 9.5, 10.775, 14, respectively. Furthermore, we give the bifurcation plot, which can be seen in Figure 4, to indicate that the bifurcation value of system (46) is 0.25.

6. Conclusions

Suppressing the chaotic behavior of nonlinear dynamical systems has been a significant and classic issue in many disciplines. For a long time, the suppression of chaos has attracted much attention from many scholars in mathematics, physics, chemistry, engineering, and numerous other areas. In the present manuscript, based on the earlier publications, we set up a novel fractional-order chaotic chemical reaction model. Taking advantage of an appropriate fractional-order PD^{ζ} controller, we can effectively eliminate the chaotic phenomenon of the involved fractional-order chaotic chemical reaction model. A delay-independent sufficient condition to guarantee the stability and the creation of Hopf bifurcation of the fractional-order controlled chaotic chemical reaction model is built. The exploration manifests that the delay occurring in fractional-order PD^{ζ} controller is the key factor in suppressing the chaotic phenomenon of the fractional-order chaotic chemical reaction approach of this manuscript are entirely new and the exploration approach of this manuscript can also be utilized to inquire into numerous chaos control problem of lots of fractional-order chaotic dynamical systems.

Data Availability

No data were used to support this study.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

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Research Article

On *q*-Convex Functions Defined by the *q*-Ruscheweyh Derivative Operator in Conic Regions

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The core objective of this article is to introduce and investigate a new class $\beta - \text{UCV}_q^{\lambda}[A, B]$ of convex functions associated with the conic domain defined by the Ruscheweyh *q*-differential operator. Many interesting properties such as sufficiency criteria, coefficient bounds, partial sums, and radius of convexity of order α for the functions of the said class are investigated here.

1. Introduction

Quantum calculus has emerged as one of the most vibrant areas of research in recent years. Researchers have discussed and found its applications in numerous dimensions, such as hypergeometric series, complex analysis, and applied physics. It has developed techniques to be used in *q*-calculus, time scales, partitions, and continued fractions. Jackson, for the first time, in the beginning of the 20th century, introduced quantum calculus, where he developed and standardized it. For more details about quantum calculus, see [1-14]. To make a good pace and understanding of the results presented in this article, we are going to give below some primary definitions and relevant details of quantum calculus. Suppose \mathcal{F} represents the class of holomorphic functions of type

$$y(z) = z + \sum_{n=2}^{\infty} b_n z^n.$$
 (1)

in open unit disk $\mathscr{C} = \{z: z \in \mathbb{C} \text{ and } |z| < 1\}$ and normalized by the conditions y'(0) = 1 and y(0) = 0. Moreover, \mathscr{S}

represents the class of all functions in \mathcal{F} which are univalent in \mathcal{E} ; see [15].

A domain \mathcal{D} is starlike with respect to a point $z_0 \in \mathcal{D}$ if all possible lines which are confined by two points, connecting z_0 to any other point, lie entirely within \mathcal{D} . Correspondingly, a domain \mathcal{C} is convex if all possible lines which are obtained by connecting any two points in \mathcal{D} lie thoroughly within \mathcal{D} . More clearly, we can say that if the domain is starlike with respect to each of its points in \mathcal{D} , then it is convex. If $y(\mathcal{C})$ is starlike for $y \in \mathcal{S}$ with respect to the origin, then it is called a starlike function, whereas if $y(\mathcal{C})$ is convex, then it is called a convex function. The class of all convex functions is represented by C, and the class of all starlike functions is represented by S^* . Analytically, these are defined as follows:

$$S^*: = \left\{ y \in \mathcal{S}: \Re\left\{\frac{zy'(z)}{y(z)}\right\} > 0, \quad z \in \mathcal{C} \right\},$$

$$C: = \left\{ y \in \mathcal{S}: \Re\left\{1 + \frac{zy''(z)}{y'(z)}\right\} > 0, \quad z \in \mathcal{C} \right\}.$$
(2)

For $\alpha \in [0, 1)$, suppose that $S^*(\alpha)$ and $C(\alpha)$ are subclasses of S consisting of α -starlike functions and α -convex functions, respectively, defined analytically as follows:

$$S^{*}(\alpha) := \left\{ y \in \mathcal{S} : \Re\left\{\frac{zy'(z)}{y(z)}\right\} > \alpha, \quad z \in \mathcal{C} \right\},$$

$$C(\alpha) := \left\{ y \in \mathcal{S} : \Re\left\{1 + \frac{zy''(z)}{y'(z)}\right\} > \alpha, \quad z \in \mathcal{C} \right\}.$$

(3)

For $\alpha = 0$, the class $S^*(\alpha) \Rightarrow S^*$ and the class $C(\alpha) \Rightarrow C$. Moreover, the following two classes are closely related with their functions defined, respectively.

$$S_{\alpha}^{*}:=\left\{y\in \mathcal{S}: \left|\frac{zy'(z)}{y(z)}-1\right|<1-\alpha, \quad z\in \mathcal{C}\right\},$$

$$C_{\alpha}:=\left\{y\in \mathcal{S}: \left|\frac{zy''(z)}{y'(z)}\right|<1-\alpha, \quad z\in \mathcal{C}\right\}.$$
(4)

Note that $S_{\alpha}^* \subseteq S^*(\alpha)$ and $C_{\alpha} \subseteq C(\alpha)$. The k^{th} partial sum of the function *y*, denoted by y_k , is the polynomial, defined by

$$y_k(z) = z + \sum_{n=2}^k b_n z^n.$$
 (5)

Generally, lower bounds on ratios such as $\Re\{y(z)/y_k(z)\}$ or $\Re\{y_k(z)/y(z)\}$ have been found to be sharp only when k = 1, but Silverman determined sharpness $\forall n \in \mathbb{N}$; see [16, 17]. He investigated that lower bounds are strictly increasing functions of k. In the present article, by using Silverman's technique [16], we will find the function's ratio having Taylor series (1) to its sequence of partial sums $y_k(z) = z + \sum_{n=2}^{k} b_n z^n$ when the coefficients of y are sufficiently small to fulfill the necessary and sufficient condition. In more details to clarify, we will find sharp lower bounds for $y(z)/y_k(z), y'(z)/y'_k(z), y_k(z)/y(z)$, and $y'_k(z)/y'(z)$. Indeed, we will use the familiar result, i.e., $\Re\{w(z) - 1/w(z) + 1\} > 0, z \in \mathcal{C}$, if and only if $w(z) = \sum_{n=1}^{\infty} c_n z^n$ satisfies $|w(z)| \le |z|$. Unless otherwise stated, we will presume that y has form (1) and that its sequence of partial sums is represented by (5).

For $\alpha \in [0, 1)$, Ravichandran gave the sharp radius of starlike and convex functions of order α with form (1) whose Taylor series coefficients b_n satisfy the conditions $|b_2| = 2 \ d, \ d \in [0, 1]$, and $|b_n| \le n$; M or $M/n \ (M > 0)$ for $n \ge 3$.

Consider that y_1 and y_2 are holomorphic functions in \mathscr{E} with w(0) = 0 and $|w(z) \le 1|$, $\forall z \in \mathscr{E}$, so that $y_1(z) = y_2(w(z))$; y_1 will be subordinated by y_2 and denoted by $y_1 < y_2$. If y_2 is holomorphic, then $y_1 < y_2$ iff $y_1(0) = y_2(0)$ and $y_1(\mathscr{E}) \le y_2(\mathscr{E})$.

For two holomorphic functions

$$y_1(z) = \sum_{k=0}^{\infty} a_k z^k$$
 and $y_2(z) = \sum_{k=0}^{\infty} b_k z^k$ ($z \in E$), (6)

the Hadamard product of $y_1(z)$ and $y_2(z)$ is defined as

$$y_1(z) * y_2(z) = \sum_{k=0}^{\infty} a_k b_k z^k.$$
 (7)

We will define some notations and concepts of quantum calculus which are to be used in this article. All results can be found in [2, 3, 18]. For $n \in \mathbb{N}$, 0 < q < 1, we see the classical *q*-theory begins with the *q*-extension of the positive numbers. The expression

$$\lim_{q \to 1} \frac{1 - q^n}{1 - q} = n.$$
(8)

proposes that we define the q-generalization of n, which is also called the q-bracket of n, given as

$$[n,q] = [n]_q = \frac{1-q^n}{1-q},$$
(9)

and the q-generalization of the factorial which is called q-factorial given by

$$[n]_{q}! = \begin{cases} [n]_{q} [n-1]_{q} \dots [1]_{q}, & n = 1, 2, \dots, \\ 1, & n = 0. \end{cases}$$
(10)

The *q*-difference operator for $y \in \mathcal{F}$ is defined as

$$\partial_q y(z) = \frac{y(qz) - y(z)}{z(q-1)}, \quad (z \in E),$$
 (11)

and we can see that, for $n \in \mathbb{N}$ and $z \in E$,

$$\partial_{q} z^{n} = [n]_{q} z^{n-1},$$

$$\partial_{q} \left\{ \sum_{n=1}^{\infty} b_{n} z^{n} \right\} = \sum_{n=1}^{\infty} [n]_{q} b_{n} z^{n-1}.$$
(12)

For $y(z) \in \mathcal{F}$, the *q*-analogue of the Ruscheweyh differential operator is defined as

$$R_{q}^{\lambda}y(z) = \varphi(q, \lambda + 1; z) * y(z)$$

= $z + \sum_{n=2}^{\infty} \psi_{n-1}b_{n}z^{n}, (z \in E \text{ and } \lambda > -1),$ (13)

where

$$\varphi(q, \lambda + 1; z) = z + \sum_{n=2}^{\infty} \psi_{n-1} z^n,$$
 (14)

and

$$\psi_{n-1} = \frac{\Gamma_q \left(\lambda + n\right)}{\left[n-1\right]_q \Gamma_q \left(\lambda + 1\right)} = \frac{\left[\lambda + 1, q\right]_{n-1}}{\left[n-1\right]_q !}, \quad \left(\psi_0 = 1\right), \quad (15)$$

where $[\lambda + 1, q]_{n-1}$ is a Pochhammer symbol, which is defined as follows:

$$[n,q]_m = \begin{cases} 1, & n = 0, \\ [n,q][n+1,q][n+2,q][n+3,q]\dots[m+n-1,q], & n \in \mathbb{N}. \end{cases}$$
(16)

From (13), it is clear that

$$R_q^0 y(z) = y(z) \text{ and } R_q^1 y(z) = z \partial_q y(z),$$

$$R_q^m y(z) = \frac{z \partial_q^m (z^{m-1} y(z))}{[m]_q!}, \quad (m \in \mathbb{N}),$$

$$\lim_{q \to 1^-} \varphi(q, \lambda + 1; z) = \frac{z}{(1-z)^{\lambda+1}},$$
(17)

$$\lim_{q \to 1^-} R_q^{\lambda} y(z) = y(z) * \frac{z}{(1-z)^{\lambda+1}}.$$

It follows that $q \rightarrow 1^-$, and the Ruscheweyh *q*-differential operator converts into the Ruscheweyh differential operator $D^{\delta}(y(z))$; for more details, see [19]. Using (13),

$$z\partial R_q^{\lambda} y(z) = \left(1 + \frac{[\lambda]_q}{q^{\lambda}}\right) R_q^{\lambda+1} y(z) - \frac{[\lambda]_q}{q^{\lambda}} R_q^{\lambda} y(z).$$
(18)

If
$$q \longrightarrow 1^-$$
, then
 $z(R^{\lambda}y(z))' = (1+\lambda)R^{\lambda+1}y(z) - \lambda R^{\lambda}y(z).$ (19)

Definition 1. The function p(z) will lie in the class $\beta - P_q[A, B]$ if and only if

$$p(z) \prec \frac{(A(1+q)+(3-q))\tilde{p}_{\beta}(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\tilde{p}_{\beta}(z) - (B(1+q)-(3-q))}, \quad \beta \ge 0,$$
(20)

where

$$\widetilde{p}_{\beta}(z) = \begin{cases} \frac{1+z}{1-z}, & \beta = 0, \\ 1 + \frac{2}{\pi^{2}} \left(\log \frac{1+\sqrt{z}}{1-\sqrt{z}} \right)^{2}, & \beta = 1, \\ 1 + \frac{2}{\pi^{2}} \sinh^{2} \left[\left(\frac{2}{\pi} \arccos \beta \right) \arctan y \sqrt{z} \right], & 0 < \beta < 1, \\ 1 + \frac{1}{\beta^{2} - 1} \sin \left(\frac{\pi}{2R(n)} \int_{0}^{u(z)/\sqrt{t}} \frac{1}{\sqrt{1-x^{2}} \sqrt{1-(tx)^{2}}} dx \right) + \frac{1}{\beta^{2} - 1}, \quad \beta > 1. \end{cases}$$

$$(21)$$

For more details, see [20–24]. If $\tilde{p}_{\beta}(z) = 1 + \delta_{\beta}z + \cdots$, then it is shown in [25] that, from (46), one can have

$$\delta_{\beta} = \begin{cases} \frac{8 (\arccos \beta)^{2}}{\pi^{2} (1 - \beta^{2})}, & 0 \le \beta < 1, \\ \frac{8}{\pi^{2}}, & \beta = 1, \\ \frac{\pi^{2}}{4 (\beta^{2} - 1) \sqrt{t} (1 + t) R^{2}(t)}, & \beta > 1. \end{cases}$$
(22)

Definition 2. A function $y(z) \in \mathcal{F}$ will lie in the class $\beta - \text{UCV}_q[A, B], \beta \ge 0, -1 \le B < A \le 1$, if and only if

$$\Re \left[\frac{(B(1+q)-(3-q))D_q(zD_qy(z))/D_qy(z)-(A(1+q)-(3-q))}{(B(1+q)+(3-q))D_q(zD_qy(z))/D_qy(z)-(A(1+q)+(3-q))} \right]$$

$$> \beta \left| \frac{(B(1+q)-(3-q))D_q(zD_qy(z))/D_qy(z)-(A(1+q)-(3-q))}{(B(1+q)+(3-q))D_q(zD_qy(z))/D_qy(z)-(A(1+q)+(3-q))} - 1 \right|,$$
(23)

or equivalently,

$$\frac{D_q(zD_qy(z))}{D_qy(z)} \in \beta - P_q[A, B].$$
(24)

For more details about the above classes and conic domain, we refer the readers to [20, 25-28]. Using the

q-Ruscheweyh differential operator, we now define the following more general class $\beta - \text{UCV}_q^{\lambda}[A, B]$ of functions associated with the conic domain defined by Janowski functions.

Definition 3. A function $y(z) \in \mathcal{F}$ will lie in the class $\beta - \text{UCV}_q^{\lambda}[A, B], \beta \ge 0, -1 \le B < A \le 1$, if and only if

$$\Re \left[\frac{(B(1+q)-(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)+(3-q))} \right]$$

$$> \beta \left| \frac{(B(1+q)-(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)+(3-q))} - 1 \right|,$$
(25)

or equivalently,

$$\frac{\partial_q \left(z \partial_q R_q^{\lambda} y(z) \right)}{\partial_q R_q^{\lambda} y(z)} \in \beta - P[A, B].$$
(26)

The above defined class $\beta - \text{UCV}_q^{\lambda}[A, B]$ generalizes many known classes which can be obtained by setting suitable particular values to the parameters as follows.

Special cases:

- (1) $\beta UCV_{1^-}^0[A, B] = \beta UCV[A, B]$, the well-known class of β -uniformly Janowski convex functions, introduced by Noor and Malik [27]
- (2) 0 UCV⁰₁ [A, B] = C[A, B], the well-known class of Janowski convex functions, introduced by Janowski [20]
- (3) $\beta \text{UCV}_{1^{-}}^{0} [1 2\alpha, -1] = \text{KD}(\beta, \alpha)$, see [29]

Lemma 1 (see [30]). Let $g(z) = 1 + \sum_{n=1}^{\infty} c_n z^n$ be subordinate to $G(z) = 1 + \sum_{n=1}^{\infty} C_n z^n$. If G(z) is holomorphic in \mathscr{C} and $G(\mathscr{C})$ is convex, then

(4) $0 - UCV_{1-}^0 [1 - 2\alpha, -1] = C(\alpha)$, see [15]

$$\left|c_{n}\right| \leq \left|C_{1}\right|, \quad n \geq 1. \tag{27}$$

2. Main Results

Theorem 1. A function $y(z) \in \mathcal{F}$ with form (1) will lie in class $\beta - UCV_q^{\lambda}[A, B]$, $\beta \ge 0$, $-1 \le B < A \le 1$, if it satisfies the condition

$$\sum_{n=2}^{\infty} \frac{\mathbb{E}_n}{\varepsilon} |b_n| < 1,$$
(28)

where

$$\mathbb{E}_{n} = [n]_{q} \left\{ 2(3-q)(\beta+1)q[n-1]_{q} + \left| (B(1+q)+(3-q))[n]_{q} - (A(1+q)+(3-q)) \right| \right\} \psi_{n-1},$$
(29)

and

$$\varepsilon = (1+q)|B-A|. \tag{30}$$

Proof. Suppose that (28) holds; then, it is enough to show that

$$\beta \left| \frac{(B(1+q)-(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)+(3-q))} - 1 \right|$$

$$- \Re \left[\frac{(B(1+q)-(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_q (z\partial_q R_q^{\lambda} y(z))/\partial_q R_q^{\lambda} y(z) - (A(1+q)+(3-q))} - 1 \right] < 1.$$
(31)

We consider

$$\begin{split} \beta \left| \frac{(B(1+q)-(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z))/\partial_{q}R_{q}^{\lambda}y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z))/\partial_{q}R_{q}^{\lambda}y(z) - (A(1+q)-(3-q))} - 1 \right| \\ & - \Re \left[\frac{(B(1+q)-(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z))/\partial_{q}R_{q}^{\lambda}y(z) - (A(1+q)-(3-q))}{(B(1+q)+(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z))/\partial_{q}R_{q}^{\lambda}y(z) - (A(1+q)+(3-q))} - 1 \right] \\ & \leq (\beta+1) \left| \frac{(B(1+q)-(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z)) - (A(1+q)-(3-q))\partial_{q}R_{q}^{\lambda}y(z)}{(B(1+q)+(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z)) - (A(1+q)+(3-q))\partial_{q}R_{q}^{\lambda}y(z)} - 1 \right| \\ & = 2(3-q)(\beta+1) \left| \frac{\partial_{q}R_{q}^{\lambda}y(z) - \partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z)) - (A(1+q)+(3-q))\partial_{q}R_{q}^{\lambda}y(z)}{(B(1+q)+(3-q))\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z)) - (A(1+q)+(3-q))\partial_{q}R_{q}^{\lambda}y(z)} \right| \end{split}$$
(32)
$$& = 2(3-q)(\beta+1) \left| \frac{\sum_{n=2}^{\infty} (1-[n]_{q})\psi_{n-1}[n]_{q}b_{n}z^{n}}{(-A(1+q)+(3-q))} \right| \psi_{n-1}[n]_{q}b_{n}z^{n}} \right| \\ & \leq \frac{2(3-q)(\beta+1)}{(1+q)|B-A|1/\psi_{n-1}-\sum_{n=2}^{\infty} |(B(1+q)+(3-q))[n]_{q} - (A(1+q)+(3-q))|[n]_{q}}|b_{n}|}{(1+q)|B-A|1/\psi_{n-1}-\sum_{n=2}^{\infty} |(B(1+q)+(3-q))[n]_{q} - (A(1+q)+(3-q))|[n]_{q}|b_{n}|}. \end{split}$$

The last expression is bounded above by 1 if

$$2(3-q)(\beta+1)\sum_{n=2}^{\infty}q[n-1]_{q}[n]_{q}|b_{n}| < (1+q)|B-A|\frac{1}{\psi_{n-1}}$$

$$-\sum_{n=2}^{\infty} |(B(1+q)+(3-q))[n]_{q} - (A(1+q)+(3-q))|[n]_{q}|b_{n}|,$$
(33)

which reduces to

$$\sum_{n=2}^{\infty} [n]_{q} \left\{ 2(3-q)(\beta+1)q[n-1]_{q} + \left| \begin{array}{c} (B(1+q)+(3-q))[n]_{q} \\ -(A(1+q)+(3-q)) \end{array} \right| \right\} \psi_{n-1} |b_{n}| < (1+q)|B-A|.$$
(34)

This finalizes the proof. \Box

For $q \rightarrow 1^-$ and $\lambda = 0$, we have the following known result, proved in [27].

Corollary 1. A function $y(z) \in \mathcal{F}$ with form (1) will lie in class $\beta - UCV[A, B]$, $\beta \ge 0$, $-1 \le B < A \le 1$, if it satisfies the condition

$$\sum_{n=2}^{\infty} n\{2(\beta+1)(n-1) + |n(B+1) - (A+1)|\} |b_n| < |B-A|.$$
(35)

For $q \longrightarrow 1^-$, $\lambda = 0$, and $A = 1 - 2\alpha$ and B = -1, we have the following known result, proved in [29].

Corollary 2. A function $y(z) \in \mathcal{F}$ with form (1) will lie in class $KD(\beta, \alpha), \beta \ge 0$, $0 \le \alpha < 1$, if it satisfies the condition

$$\sum_{n=2}^{\infty} n\{n(\beta+1) - (\beta+\alpha)\} |b_n| < (1-\alpha).$$
(36)

Theorem 2. Let $y(z) \in \beta - UCV_q^{\lambda}[A, B], \quad \beta \ge 0,$ $-1 \le B < A \le 1$, and be of form (1); then, for $n \ge 2$,

$$\left| b_n \right| \le \frac{1}{[n]_q} \prod_{j=0}^{n-2} \frac{\left| (A-B)(q+1)\delta_\beta \psi_j - 4Bq[j]_q \psi_j \right|}{4q[j+1]_q \psi_{j+1}}, \qquad (37)$$

where ψ is defined by (15).

Proof. By the definition for $y(z) \in \beta - UCV_q^{\lambda}[A, B]$, we have

$$\frac{\partial_q \left(z \partial_q R_q^{\lambda} y(z) \right)}{\partial_q R_q^{\lambda} y(z)} = p(z), \tag{38}$$

where

$$p(z) \prec \frac{(A(1+q)+(3-q))\tilde{p}_{\beta}(z)-(A(1+q)-(3-q))}{(B(1+q)+(3-q))\tilde{p}_{\beta}(z)-(B(1+q)-(3-q))}.$$
(39)

If $\tilde{p}_{\beta}(z) = 1 + \delta_{\beta}z + \dots$, then

$$\frac{(A(1+q)+(3-q))\tilde{p}_{\beta}(z)-(A(1+q)-(3-q))}{(B(1+q)+(3-q))\tilde{p}_{\beta}(z)-(B(1+q)-(3-q))} = 1 + \frac{1}{4}(A-B)(q+1)\delta_{\beta} + \frac{1}{4}\left[\left(-\frac{1}{4}Aq - \frac{1}{4}A + \frac{1}{4}Bq + \frac{1}{4}B\right)((B+1)(1+q)+2-2q)\right]\delta_{\beta}^{2} + \cdots$$
(40)

Now, if $p(z) = 1 + \sum_{n=1}^{\infty} c_n z^n$, then by (27) and (39), we have

$$|c_n| \le \frac{1}{4} (A - B) (q + 1) |\delta_\beta|, \quad n \ge 1.$$
 (41)

Now, from (38), we have

$$\partial_q \left(z \partial_q R_q^{\lambda} y(z) \right) = p(z) \partial_q R_q^{\lambda} y(z).$$
(42)

Let $p(z) = 1 + \sum_{n=1}^{\infty} c_n z^n$, and using the Cauchy product formula, we obtain

$$1 + \sum_{n=2}^{\infty} [n]_{q} [n]_{q} \psi_{n-1} b_{n} z^{n-1} = \left(1 + \sum_{n=1}^{\infty} c_{n} z^{n}\right) \left(1 + \sum_{n=2}^{\infty} [n]_{q} \psi_{n-1} b_{n} z^{n-1}\right)$$

$$\sum_{n=2}^{\infty} [n]_{q} [n]_{q} \psi_{n-1} b_{n} z^{n-1} = \sum_{n=2}^{\infty} c_{n-1} z^{n-1} + \sum_{n=2}^{\infty} [n]_{q} \psi_{n-1} b_{n} z^{n-1} + \left(\sum_{n=2}^{\infty} [n]_{q} \psi_{n-1} b_{n} z^{n-1}\right) \cdot \left(\sum_{n=2}^{\infty} c_{n-1} z^{n-1}\right).$$
(43)

This implies that

$$\sum_{n=2}^{\infty} ([n]_q - 1)[n]_q \psi_{n-1} b_n z^{n-1} = \left(\sum_{n=2}^{\infty} c_{n-1} + \sum_{n=2}^{\infty} \sum_{j=2}^{n} [j]_q \psi_{j-1} b_j c_{n-j} \right) z^{n-1}.$$
(44)

Comparison of coefficients of z^{n-1} gives us

$$\left([n]_{q}-1\right)[n]_{q}\psi_{n-1}b_{n}=\left(c_{n-1}+\sum_{j=2}^{n}[j,q]\psi_{j-1}b_{j}c_{n-j}\right)(b_{1}=1),\quad(45)$$

or

$$b_{n} = \frac{1}{\left([n]_{q} - 1\right)[n]_{q}\psi_{n-1}} \left(c_{n-1} + \sum_{j=2}^{n} [j]_{q}\psi_{j-1}b_{j}c_{n-j}\right).$$
(46)

Using (41), we have

$$\left|b_{n}\right| \leq \frac{(A-B)\left(q+1\right)\left|\delta_{\beta}\right|}{4q[n-1]_{q}[n]_{q}\psi_{n-1}}\left(1+\sum_{j=2}^{n-1}\left[j\right]_{q}\psi_{j-1}\left|b_{j}\right|\right), \quad (47)$$

or

$$\left|b_{n}\right| \leq \frac{(A-B)(q+1)\left|\delta_{\beta}\right|}{4q[n-1]_{q}[n]_{q}\psi_{n-1}} \left(\sum_{j=1}^{n-1} [j]_{q}\psi_{j-1}\left|b_{j}\right|\right).$$
(48)

Now, we prove that

$$\frac{(A-B)(q+1)\left|\delta_{\beta}\right|}{4q[n-1]_{q}[n]_{q}\psi_{n-1}}\left(\sum_{j=1}^{n-1}[j]_{q}\psi_{j-1}\left|b_{j}\right|\right) \leq \frac{1}{[n]_{q}}\prod_{j=0}^{n-2}\frac{\left|(A-B)(q+1)\delta_{\beta}\psi_{j}-4Bq[n]_{q}\psi_{j}\right|}{4q[j+1]_{q}\psi_{j+1}}.$$
(49)

or

For this, we use the induction technique. For n = 2, we have from (46),

$$|b_2| \le \frac{\left|\delta_{\beta}\right| (q+1) (A-B)}{4q[1]_q [2]_q \psi_1} \sum_{j=1}^{2^{-1}} [j]_q \psi_{j-1} |b_j|, \qquad (50)$$

$$|b_2| \le \frac{\left|\delta_\beta\right| (A - B)(q + 1)}{4q[1]_q[2]_q \psi_1}, \quad \psi_0 = 1.$$
(51)

For n = 3, we have from (46),

$$\begin{aligned} |b_{3}| &\leq \frac{\left|\delta_{\beta}\right| (A-B) (q+1)}{4q[2]_{q}[3]_{q}\psi_{2}} \sum_{j=1}^{2} [j]_{q}\psi_{j-1} |b_{j}| \\ &= \frac{\left|\delta_{\beta}\right| (A-B) (q+1)}{4q[2]_{q}[3]_{q}\psi_{2}} \left([1]_{q}\psi_{0} |b_{1}| + [2]_{q}\psi_{1} |b_{2}| \right) \\ &\leq \frac{\left|\delta_{\beta}\right| (A-B) (q+1)}{4q[2]_{q}[3]_{q}\psi_{2}} \left(1 + \frac{(A-B) (q+1) |\delta_{\beta}|}{4q[1]_{q}} \right). \end{aligned}$$
(52)

From (37), we have

$$\begin{aligned} |b_{3}| &\leq \frac{1}{[3]_{q}} \prod_{j=0}^{1} \frac{\left| (A-B) \left(q+1\right) \delta_{\beta} \psi_{j} - 4Bq[j]_{q} \psi_{j} \right|}{4q[j+1]_{q} \psi_{j+1}} \\ &= \frac{1}{[3]_{q}} \frac{(A-B) \left(q+1\right) \left| \delta_{\beta} \right|}{4q[1]_{q} \psi_{1}} \left(\frac{(A-B) \left(q+1\right) \left| \delta_{\beta} \right| \psi_{1} + 4q[1]_{q} \psi_{1}}{4q[2]_{q} \psi_{2}} \right) \\ &= \frac{\left| \delta_{\beta} \right| (A-B) \left(q+1\right)}{4q[2]_{q} [3]_{q} \psi_{2}} \left(1 + \frac{(A-B) \left(q+1\right) \left| \delta_{\beta} \right|}{4q[1]_{q}} \right). \end{aligned}$$
(53)

Let the assumption be true for n = m + 1. From (46), we From (37), we have have

$$\left|b_{m}\right| \leq \frac{\left|\delta_{\beta}\right| (A-B) (q+1)}{4q[m-1]_{q}[m]_{q}\psi_{m-1}} \left(\sum_{j=1}^{m-1} [j]_{q}\psi_{j-1} \left|b_{j}\right|\right).$$
(54)

$$\left|b_{m}\right| \leq \frac{1}{[m]_{q}} \prod_{j=0}^{m-2} \frac{\left|(A-B)\left(q+1\right)\delta_{\beta}\psi_{j}-4Bq[j]_{q}\psi_{j}\right|}{4q[j+1]_{q}\psi_{j+1}}.$$
 (55)

By the induction hypothesis,

$$\frac{1}{[m]_{q}}\prod_{j=0}^{m-2}\frac{\left|(A-B)\left(q+1\right)\delta_{\beta}\psi_{j}-4Bq[j]_{q}\psi_{j}\right|}{4q[j+1]_{q}\psi_{j+1}} \ge \frac{\left|\delta_{\beta}\right|(A-B)\left(q+1\right)}{4q[m-1]_{q}[m]_{q}\psi_{m-1}}\sum_{j=1}^{m-1}[j]_{q}\psi_{j-1}\left|b_{j}\right|.$$
(56)

Multiplying both sides by $1/[m]_q (A-B)(q+1)|\delta_\beta|\psi_{m-1}+4q[m-1]_q\psi_{m-1}/4q[m]_q\psi_m$, we have

$$\begin{split} \frac{1}{[m]_{q}} \prod_{j=0}^{m-2} \frac{\left|(A-B)(q+1)\delta_{\beta}\psi_{j}-4Bq[j]_{q}\psi_{j}\right|}{4q[j+1]_{q}\psi_{j+1}} \\ &\geq \left(\frac{1}{[m]_{q}} \frac{(A-B)(q+1)\left|\delta_{\beta}\right|\psi_{m-1}+4q[m-1]_{q}\psi_{m-1}}{4q[m]_{q}\psi_{m}}\right) \times \\ &\cdot \left(\frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m-1]_{q}\psi_{m-1}} \sum_{j=1}^{m-1}\psi_{j-1}\left|b_{j}\right|\right) \\ &= \frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m]_{q}\psi_{m}} \times \\ &\cdot \left(\psi_{m-1}\frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m-1]_{q}[m]_{q}\psi_{m-1}} \sum_{j=1}^{m-1}\psi_{j-1}\left|b_{j}\right| + \frac{1}{[m]_{q}} \sum_{j=1}^{m-1}\psi_{j-1}\left|b_{j}\right|\right) \\ &\geq \frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m]_{q}\psi_{m}} \left(\psi_{m-1}\left|b_{m}\right| + \frac{1}{[m]_{q}} \sum_{j=1}^{m-1}\psi_{j-1}\left|b_{j}\right|\right) \\ &= \frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m]_{q}\psi_{m}} \sum_{j=1}^{m}\psi_{j-1}\left|b_{j}\right|. \end{split}$$
(57)

That is,

$$\frac{\left|\delta_{\beta}\right|(A-B)(q+1)}{4q[m-1]_{q}[m]_{q}\psi_{m-1}}\sum_{j=1}^{m-1}\psi_{j-1}\left|b_{j}\right| \leq \frac{1}{[m]_{q}}\prod_{j=0}^{m-2}\frac{\left|(A-B)(q+1)\delta_{\beta}\psi_{j}-4Bq[j]_{q}\psi_{j}\right|}{4q[j+1]_{q}\psi_{j+1}}.$$
(58)

Hence, the consequence is true for n = m + 1. Therefore, using mathematical induction, we have proved that (37) is true $\forall n, n \ge 2$. \Box

For $q \rightarrow 1^-$ and $\lambda = 0$, we have the following known result, proved in [27].

Corollary 3. Let $y(z) \in \beta - UCV[A, B]$, $\beta \ge 0$, $-1 \le B < A \le 1$, and be of form (1); then, for $n \ge 2$,

$$|b_n| \le \frac{1}{n} \prod_{j=0}^{n-2} \frac{|(A-B)\delta_\beta - 2Bj|}{2(j+1)}.$$
 (59)

For $q \longrightarrow 1^-$, $\lambda = 0$, and $A = 1 - 2\alpha$ and B = -1, we have the following known result, proved in [29].

Corollary 4. Let $y(z) \in KD(\beta, \alpha)$, $\beta \ge 0$, $0 \le \alpha < 1$, and be of form (1); then, for $n \ge 2$,

$$|b_n| \le \frac{1}{n} \prod_{j=0}^{n-2} \frac{\left| (1-\alpha)\delta_\beta + j \right|}{(j+1)}.$$
 (60)

Using the already proven results of Silverman [16] and Silvia [17] on partial sums of holomorphic functions, we will find the fraction of (1) to its sequence of partial sums $y_k(z) =$ $z + \sum_{n=2}^k b_n z^n$ when the function y(z) has coefficients small enough to satisfy condition (28). We will investigate sharp lower bounds for $\Re\{y(z)/y_k(z)\}$, $\Re\{y'(z)/y'_k(z)\}$, $\Re\{y_k(z)/y(z)\}$, and $\Re\{y'_k(z)/y'(z)\}$ in the class $\beta - \text{UCV}_a^{\lambda}[A, B]$.

Theorem 3. If $y(z) \in \beta - UCV_q^{\lambda}[A, B]$, then

$$\Re\left\{\frac{y(z)}{y_k(z)}\right\} \ge 1 - \frac{\varepsilon}{\mathbb{E}_{k+1}},\tag{61}$$

where \mathbb{E}_{k+1} is defined by (29) and $\varepsilon = (1+q)|B-A|$. The extremal function

$$y(z) = z + \frac{\varepsilon}{\mathbb{E}_{k+1}} z^{k+1}.$$
 (62)

gives the sharp result.

Proof. Define a function w(z):

$$w(z) = \frac{\mathbb{E}_{k+1}}{\varepsilon} \left[\frac{y(z)}{y_k(z)} - \left(1 - \frac{\varepsilon}{\mathbb{E}_{k+1}} \right) \right], \tag{63}$$

and this will reduce to

$$= \frac{\mathbb{E}_{k+1} \left(1 + \sum_{n=2}^{\infty} b_n z^{n-1} \right)}{\varepsilon \left(1 + \sum_{n=2}^{k} b_n z^{n-1} \right)} - \frac{\mathbb{E}_{k+1}}{\varepsilon} + 1$$

$$(54)$$

$$(z) = \frac{1 + \sum_{n=2}^{k} b_n z^{n-1} + \mathbb{E}_{k+1} / \varepsilon \sum_{n=k+1}^{\infty} b_n z^{n-1}}{1 + \sum_{n=2}^{k} b_n z^{n-1}}.$$

We have

w

$$\frac{w(z)-1}{w(z)+1} \le \frac{\mathbb{E}_{k+1}/\varepsilon \sum_{n=k+1}^{\infty} |b_n|}{2-2\sum_{n=2}^{k} |b_n| - \mathbb{E}_{k+1}/\varepsilon \sum_{n=k+1}^{\infty} |b_n|}.$$
(65)

Now,

$$\left|\frac{w(z)-1}{w(z)+1}\right| \le 1 \tag{66}$$

if

$$\sum_{n=2}^{k} \left| b_n \right| + \frac{\mathbb{E}_{k+1}}{\varepsilon} \sum_{n=k+1}^{\infty} \left| b_n \right| \le 1.$$
(67)

It is sufficient to show that the left hand side of (28) is bounded above by $\sum_{n=2}^{\infty} \mathbb{E}_n / \varepsilon |b_n|$ if

$$\sum_{n=2}^{k} \left| b_n \right| + \frac{\mathbb{E}_{k+1}}{\varepsilon} \sum_{n=k+1}^{\infty} \left| b_n \right| \le \sum_{n=2}^{\infty} \frac{\mathbb{E}_n}{\varepsilon} \left| b_n \right|.$$
(68)

This leads to the following expression:

$$\sum_{n=2}^{k} \left(\frac{\mathbb{E}_{n} - \varepsilon}{\varepsilon}\right) \left|b_{n}\right| + \left(\frac{\mathbb{E}_{n} - \mathbb{E}_{k+1}}{\varepsilon}\right) \sum_{n=k+1}^{\infty} \left|b_{n}\right| \ge 0.$$
(69)

To ensure that the function defined by (62) gives the sharp outcome, we note that, for $z = re^{i\pi/n}$,

$$\frac{y(z)}{y_k(z)} = 1 + \frac{\varepsilon}{\mathbb{E}_{k+1}} z^n$$

$$= 1 + \frac{\varepsilon}{\mathbb{E}_{k+1}} r^n e^{\frac{i\pi}{n}}$$

$$= 1 + \frac{\varepsilon r^n}{\mathbb{E}_{k+1}} \left(\cos\frac{\pi}{n} + i\sin\frac{\pi}{n}\right)$$
(70)
$$= 1 - \frac{\varepsilon r^n}{\mathbb{E}_{k+1}}$$

$$\frac{y(z)}{y_k(z)} = \frac{\mathbb{E}_{k+1} - \varepsilon}{\mathbb{E}_{k+1}} \text{ when } r \longrightarrow 1.$$

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Theorem 4. If $y(z) \in \beta - UCV_q^{\lambda}[A, B]$, then

$$\Re\left\{\frac{y_k(z)}{y(z)}\right\} \ge \frac{\mathbb{E}_{k+1}}{\mathbb{E}_{k+1} + \varepsilon},\tag{71}$$

where \mathbb{E}_{k+1} is defined by (29) and $\varepsilon = (1+q)|B-A|$. The result (71) is sharp with the function given by (62).

Proof. Define the function w(z):

$$w(z) = \frac{\mathbb{E}_{k+1} + \varepsilon}{\varepsilon} \left[\frac{y_k(z)}{y(z)} - \frac{\mathbb{E}_{k+1}}{\mathbb{E}_{k+1} + \varepsilon} \right]$$

$$= \frac{1 + \sum_{n=2}^k b_n z^{n-1} - \mathbb{E}_{k+1}/\varepsilon \sum_{n=k+1}^\infty b_n z^{n-1}}{1 + \sum_{n=2}^\infty b_n z^{n-1}}.$$
(72)

This will become

$$\frac{w(z)-1}{w(z)+1} = \frac{\sum_{n=2}^{k} b_n z^{n-1} - \sum_{n=2}^{\infty} b_n z^{n-1} - \mathbb{E}_{k+1} / \varepsilon \sum_{n=k+1}^{\infty} b_n z^{n-1}}{2 + \sum_{n=2}^{k} b_n z^{n-1} + \sum_{n=2}^{\infty} b_n z^{n-1} - \mathbb{E}_{k+1} / \varepsilon \sum_{n=k+1}^{\infty} |b_n| z^{k-1}}$$

$$= \frac{-(1 + \mathbb{E}_{k+1} / \varepsilon) \sum_{n=k+1}^{\infty} b_n z^{n-1}}{2 + 2 \sum_{n=2}^{k} b_n z^{n-1} + (1 - \mathbb{E}_{k+1} / \varepsilon) \sum_{n=k+1}^{\infty} |b_n| z^{k-1}}.$$
(73)

This implies that

$$\left|\frac{w(z)-1}{w(z)+1}\right| \le \frac{\left(1 + \mathbb{E}_{k+1}/\varepsilon\right)\sum_{n=k+1}^{\infty} |b_n|}{2 - 2\sum_{n=2}^{k} |b_n| - \left(1 - \mathbb{E}_{k+1}/\varepsilon\right)\sum_{n=k+1}^{\infty} |b_n|}.$$
 (74)

Now,

$$\left|\frac{w(z) - 1}{w(z) + 1}\right| \le 1 \tag{75}$$

if

$$\sum_{n=2}^{k} \left| b_n \right| + \sum_{n=k+1}^{\infty} \left| b_n \right| \le 1.$$
(76)

It would be enough to show that the left side of (28) is bounded above by $\sum_{n=2}^{\infty} \mathbb{E}_n / \varepsilon |b_n|$ if

$$\sum_{n=2}^{k} \left| b_n \right| + \sum_{n=k+1}^{\infty} \left| b_n \right| \le \sum_{n=2}^{\infty} \frac{\mathbb{E}_n}{\varepsilon} \left| b_n \right|,\tag{77}$$

which leads to the following expression:

$$\sum_{n=2}^{k} \left(\frac{\mathbb{E}_{n}}{\varepsilon} - 1\right) \left| b_{n} \right| + \sum_{n=k+1}^{\infty} \left(\frac{\mathbb{E}_{n}}{\varepsilon} - 1\right) \left| b_{n} \right| \ge 0$$
(78)

or

$$\sum_{n=2}^{\infty} \left(\frac{\mathbb{E}_n}{\varepsilon} - 1\right) |b_n| \ge 0.$$
(79)

Consequently, the equality holds for the extreme function y(z) given by (62). \Box

We now turn to fractions related to the derivatives.

Theorem 5. If
$$y(z) \in \beta - UCV_q^{\lambda}[A, B]$$
, then

$$\Re\left\{\frac{y'(z)}{y'_k(z)}\right\} \ge \frac{\mathbb{E}_{k+1} - \varepsilon(k+1)}{\mathbb{E}_{k+1}},$$
(80)

where \mathbb{E}_{k+1} is defined by (29) and $\varepsilon = (1+q)|B-A|$. The result (80) is sharp with the function given by (62).

Proof. Define the function w(z):

$$w(z) = \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1)} \cdot \left[\frac{y'(z)}{y'_{k}(z)} - \frac{\mathbb{E}_{k+1} - \varepsilon(k+1)}{\mathbb{E}_{k+1}} \right]$$
$$= \frac{\mathbb{E}_{k+1} \left(1 + \sum_{n=2}^{\infty} nb_{n} z^{n-1} \right)}{\varepsilon(k+1) \left(1 + \sum_{n=2}^{k} nb_{n} z^{n-1} \right)} - \frac{\left(\mathbb{E}_{k+1} - \varepsilon(k+1)\right)}{\varepsilon(k+1)},$$
(81)

and this will reduce to

$$w(z) = \frac{1 + \sum_{n=2}^{k} nb_n z^{n-1} + \mathbb{E}_{k+1} / \varepsilon(k+1) \sum_{n=k+1}^{\infty} nb_n z^{n-1}}{1 + \sum_{n=2}^{k} nb_n z^{n-1}}.$$
(82)

Now, we have

$$\frac{w(z)-1}{w(z)+1} = \frac{\mathbb{E}_{k+1}/\varepsilon(k+1)\sum_{n=k+1}^{\infty}nb_n z^{n-1}}{2+2\sum_{n=2}^k nb_n z^{n-1} + \mathbb{E}_{k+1}/\varepsilon(k+1)\sum_{n=k+1}^{\infty}nb_n z^{n-1}}.$$
(83)

This implies that

$$\left|\frac{w(z)-1}{w(z)+1}\right| \le \frac{\mathbb{E}_{k+1}/\varepsilon(k+1)\sum_{n=k+1}^{\infty} n|b_n|}{2-2\sum_{n=2}^k n|b_n| - \mathbb{E}_{k+1}/\varepsilon(k+1)\sum_{n=k+1}^{\infty} n|b_n|}.$$
 (84)

Now,

$$\left|\frac{w(z) - 1}{w(z) + 1}\right| \le 1$$
(85)

if

$$\sum_{n=2}^{k} n \left| b_n \right| + \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1)} \sum_{k=n+1}^{\infty} n \left| b_n \right| \le 1.$$
(86)

It would be enough to show that the left side of (28) is bounded above by $\sum_{n=2}^{\infty} \mathbb{E}_n / \varepsilon |b_n|$ if

$$\sum_{n=2}^{k} n|b_n| + \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1)} \sum_{n=k+1}^{\infty} n|b_n| \le \sum_{n=2}^{\infty} \frac{\mathbb{E}_k}{\varepsilon} |b_n|, \qquad (87)$$

which leads to the following expression:

$$\sum_{n=2}^{k} \left(\frac{\mathbb{E}_{n}}{\varepsilon} - n\right) \left| b_{n} \right| + \sum_{n=k+1}^{\infty} \left(\frac{\mathbb{E}_{n}}{\varepsilon} - \frac{n\mathbb{E}_{k+1}}{\varepsilon(k+1)}\right) \left| b_{n} \right| \ge 0.$$
(88)

The result (80) is sharp with respect to the function given by (62). \Box

Theorem 6. If
$$y(z) \in \beta - UCV_a^{\lambda}[A, B]$$
, then

$$\Re\left\{\frac{y_k'(z)}{y'(z)}\right\} \ge \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1) + \mathbb{E}_{k+1}},\tag{89}$$

where \mathbb{E}_{k+1} is defined by (29) and $\varepsilon = (1+q)|B-A|$. The result (89) is sharp with respect to the function given by (62).

Proof. Define the function w(z):

$$w(z) = \frac{\varepsilon(k+1) + \mathbb{E}_{k+1}}{\varepsilon(k+1)} \cdot \left[\frac{y'_k(z)}{y'(z)} - \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1) + \mathbb{E}_{k+1}} \right]$$
$$= \frac{\left(\varepsilon(k+1) + \mathbb{E}_{k+1}\right) \left(1 + \sum_{n=2}^k nb_n z^{n-1}\right)}{\varepsilon(k+1) \left(1 + \sum_{n=2}^\infty nb_n z^{n-1}\right)} - \frac{\mathbb{E}_{k+1}}{\varepsilon(k+1)}.$$
(90)

This will become

$$w(z) = \frac{1 + \sum_{n=2}^{k} nb_n z^{n-1} - \mathbb{E}_{k+1} / \varepsilon(k+1) \sum_{n=k+1}^{\infty} nb_n z^{n-1}}{\left(1 + \sum_{n=2}^{\infty} nb_n z^{n-1}\right)}.$$
(91)

This leads us to

$$\frac{w(z)-1}{w(z)+1} = \frac{-\sum_{n=k+1}^{\infty} (1 + \mathbb{E}_{k+1}/\varepsilon(k+1)) n b_n z^{n-1}}{2 + 2\sum_{n=2}^k n b_n z^{n-1} + \sum_{n=k+1}^{\infty} (1 - \mathbb{E}_{k+1}/\varepsilon(k+1)) n b_n z^{n-1}},$$
(92)

 $\left|\frac{w(z)-1}{w(z)+1}\right| \le \frac{\left(1 + \mathbb{E}_{k+1}/\varepsilon(k+1)\right)\sum_{n=k+1}^{\infty} n|b_n|}{2 - 2\sum_{n=2}^{k} n|b_n| - \left(1 - \mathbb{E}_{k+1}/\varepsilon(k+1)\right)\sum_{n=k+1}^{\infty} n|b_n|}$

which reduces to

Now,

$$\left|\frac{w(z) - 1}{w(z) + 1}\right| \le 1.$$
(94)

if

$$\sum_{n=2}^{k} n |b_n| + \sum_{n=k+1}^{\infty} n |b_n| \le 1.$$
(95)

It is sufficient to show that the left hand side of (28) is bounded above by $\sum_{n=2}^{\infty} \mathbb{E}_n / \varepsilon |b_n|$ if

$$\sum_{n=2}^{k} n |b_n| + \sum_{n=k+1}^{\infty} n |b_n| \le \sum_{n=2}^{\infty} \frac{\mathbb{E}_n}{\varepsilon} |b_n|, \tag{96}$$

which leads to the following expression:

$$\sum_{n=2}^{\infty} \left(\frac{\mathbb{E}_n}{\varepsilon} - n\right) |b_n| \ge 0.$$
(97)

The result (89) is sharp with respect to the function given by (62). \Box

In the next theorem, we will find the radii of starlikeness for the class $\beta - \text{UCV}_q^{\lambda}[A, B]$.

Theorem 7. Let $y(z) \in \beta - UCV_q^{\lambda}[A, B]$. Then, y(z) is a convex function of order $\alpha \in [0, 1)$ in $|z| < r = r_1(\alpha)$, where

$$r_1(\alpha) = \left(\frac{\mathbb{E}_n(1-\alpha)}{\varepsilon(q[n-1]_q+(1-\alpha))}\right)^{1/n-1}, \quad n = 2, 3, \dots,$$
(98)

where \mathbb{E}_n is defined by (29) and $\varepsilon = (1+q)|B-A|$.

Proof. Let $y(z) \in \beta - \text{UCV}_q^{\lambda}[A, B]$. Then, by the theorem,

(93)
(100)

$$\sum_{n=2}^{\infty} \frac{\mathbb{E}_n}{\varepsilon} \left| b_n \right| < 1, \tag{99}$$

where \mathbb{E}_n is defined by (29) and $\varepsilon = (1+q)|B-A|$. For $\alpha \in [0, 1)$, we need to show that

that is,

$$\left| \frac{\partial_{q} \left(z \partial_{q} R_{q}^{\lambda} y(z) \right) - \partial_{q} R_{q}^{\lambda} y(z)}{\partial_{q} R_{q}^{\lambda} y(z)} \right| = \left| -\frac{\sum_{n=2}^{\infty} q[n-1]_{q} [n]_{q} \psi_{n-1} b_{n} z^{n-1}}{1 - \sum_{n=2}^{\infty} [n]_{q} \psi_{n-1} b_{n} z^{n-1}} \right| \\
\leq \frac{\sum_{n=2}^{\infty} q[n-1]_{q} [n]_{q} \psi_{n-1} |b_{n}| |z|^{n-1}}{1 - \sum_{n=2}^{\infty} [n]_{q} \psi_{n-1} |b_{n}| |z|^{n-1}}$$
(101)

Thus,
$$\left|\partial_{q}(z\partial_{q}R_{q}^{\lambda}y(z)) - \partial_{q}R_{q}^{\lambda}y(z)/\partial_{q}R_{q}^{\lambda}y(z)\right| \leq 1 - \alpha$$
 if
 $\left(\frac{q[n-1]_{q}}{1-\alpha} + 1\right)[n]_{q}\psi_{n-1}\left|b_{n}\right|\left|z\right|^{n-1} \leq 1.$ (102)

According to theorem (99), inequality (102) will be true

$$\left(\frac{q[n-1]_q}{1-\alpha}+1\right)|z|^{n-1} \le \frac{\mathbb{E}_n}{\varepsilon}.$$
(103)

Solving (103) for |z|, we obtain

$$|z|^{n-1} \le \frac{\mathbb{E}_n (1-\alpha)}{\varepsilon \left(q[n-1]_q + (1-\alpha)\right)}.$$
(104)

Setting $|z| = r(\alpha)$ in (104), we have

$$r(\alpha) = \left(\frac{\mathbb{E}_n(1-\alpha)}{\varepsilon(q[n-1]_q + (1-\alpha))}\right)^{1/n-1},$$
 (105)

which is the required result. \Box

3. Conclusion

In this article, we have applied the *q*-Ruscheweyh differential operator to define and study a new class $\beta - \text{UCV}_q^{\lambda}[A, B]$ of *q*-convex functions associated with the conic domain. This class generalizes the classes $\beta - \text{UCV}[A, B]$, C[A, B], $K(\beta, \alpha)$, $C(\alpha)$, and C which have been defined and studied earlier. This fact has been illustrated above with details and proper referencing. The results presented include sufficiency criteria related to Taylor series coefficients, the coefficient bounds, and the ratios of partial sums to their infinite sum for functions of the class $\beta - \text{UCV}_q^{\lambda}[A, B]$.

Data Availability

No data were used in this article.

Conflicts of Interest

The authors declare that there are no conflicts of interest regarding the publication of this article.

 $\left|\frac{\partial_q \left(z \partial_q R_q^{\lambda} y(z)\right)}{\partial_a R_a^{\lambda} y(z)}\right| < 1 - \alpha,$

Authors' Contributions

All authors contributed equally and approved the final manuscript.

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Research Article

Analysis of the Fractional-Order Kaup–Kupershmidt Equation via Novel Transforms

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In this article, we develop a technique to determine the analytical result of some Kaup–Kupershmidt equations with the aid of a modified technique called the new iteration transform method. This technique is a mixture of the novel integral transformation Elzaki transformation and the new iteration technique. The nonlinear term can be handled easily by a new iteration technique. The results show that the combination of the Elzaki transformation and the new iteration technique is quite capable and basically well suited for applying in such problems and that it can be implemented to other nonlinear models. This technique is viewed as an effective alternative approach to certain existing approaches for such accurate models.

1. Introduction

Fractional calculus is regarded as an important branch of science, particularly for phenomena that cannot be defined by basic nonlinear ordinary differential equations or partial differential equations with integer-order operators. The use of memory is one of the main advantages of fractional-order derivatives over standard derivatives. In recent years, there have been numerous applications of fractional-order ordinary and partial differential equations in many fields of physics and engineering. There have been several key works discovered, particularly in genetic mechanics and in the viscoelasticity concept, where fractional-order derivatives are utilized for a good explanation of the properties of materials. This is the main benefit of fractional derivatives compared with traditional integer-order models in which such effects are neglected. The computational modeling and analysis of structures and procedures, based on the explanation of their properties in concepts of fractional derivatives, obviously result in differential equations of fractional

order and the requirement of finding solutions such as mathematical equations [1-10].

The fractional-order Kaup-Kupershmidt equation is used to investigate the analysis of capillary gravity waves' attitude and nonlinear dispersive waves. The extensive fifthorder nonlinear development equation is written as

$$D^{\rho}_{\tau}\mu(\zeta,\tau) + \alpha\mu\mu_{\zeta\zeta\zeta} + \beta p\mu_{\zeta}\mu_{\zeta\zeta} + \gamma\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} = 0, \quad (1)$$

with the initial condition

$$\mu(\zeta, 0) = g(\zeta), \tag{2}$$

where α , β , and γ are real constants and $0 < \rho \le 1$ is the parameter symbolizing the order of the fractional-order derivative. By considering different values for α , β , and γ , the overload nonlinear fifth-order development model can be scaled down to the fifth-order fractional-order Kaup–Kupershmidt equation.

For $\alpha = -15$, $\beta = -15$, and $\gamma = 45$, the above equation simplifies to

$$D^{\rho}_{\tau}\mu(\zeta,\tau) - 15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} = 0,$$
(3)

with the initial condition

$$\mu(\zeta, 0) = g(\zeta). \tag{4}$$

In 1980, Kaup [11] first introduced a significant dispersive basic Kaup-Kupershmidt equation, and then it was improved by Kupershmidt [12] in 1994. This study is concerned with the analysis of the modified fractionalorder Kaup-Kupershmidt (KK) equation. In recent decades, excellent scientific work has been devoted to the analysis of the classical KK equation. The modern KK equation can be integrated at p = 5/2 [13] and is considered to have bilinear representation [14]. Soliton and solitary wave results can be obtained for general nonlinear development problems by importing four diverse techniques autonomously. Nonlaopon et al. [15] used the inverse scattering approach to establish soliton results to analyze nonlinear equations with physical implications. Two integrable differential-difference equations exhibit soliton solutions of the Kaup-Kupershmidt equation type [16]. Musette introduced the fifth-order KK equation, and Verhoeven was one of the combined instances of the Henon-Heiles method; see [17] for more details. Prakasha et al. [18] used the q-homotopy analysis transform method which is implemented to obtain the result for the fractionalorder KK equation.

Daftardar-Gejji and Jafari [19] introduced a new iterative methodology for investigating nonlinear equations in 2006. Jafari [20] was the first to use the Laplace transform in an iterative technique. In [21], Jafari et al. suggested a modified straightforward methodology, named iterative Laplace transformation technique, to look for the numerical effects of the fractional partial differential equation system. Iterative Laplace transformation technique is used to solve linear and nonlinear partial differential equations such as time-fractional Zakharov–Kuznetsov equation [22], fractional-order Fokker–Planck equation [23], and Fornberg–Whitham equation [24].

This article modified the iterative method with the Elzaki transform; the novel approach is named the iterative transformation technique. The new iterative transformation technique is implemented to evaluate the fractional order of the system of the KK equation. The outcome of several illustrative cases is described to demonstrate the effectiveness of the proposed technique. The present method is used to obtain the results of fractional-order and integral-order models. The new method reduces computing costs while increasing rate convergence. The proposed method is also helpful in dealing with other fractional-order linear and nonlinear partial differential equations.

2. Basic Definitions

Definition 1 (see [25–27]). The fractional-order Riemann–Liouville operator D^{ρ} of order ρ is defined as

$$D^{\rho}\nu(\zeta) = \begin{cases} \frac{d^{r}}{d\zeta^{\kappa}}\nu(\zeta), & \text{for } \rho = \kappa, \\ \\ \frac{1}{\Gamma(\kappa-\rho)}\frac{d}{d\zeta^{\kappa}}\int_{0}^{\zeta}\frac{\nu(\zeta)}{(\zeta-\psi)^{\rho-\kappa+1}}\mathrm{d}\psi, & \text{for } \kappa-1 < \rho < \kappa, \end{cases}$$
(5)

where $\kappa \in \mathbb{Z}^+$, $\rho \in \mathbb{R}^+$, and

$$D^{-\rho}\nu(\zeta) = \frac{1}{\Gamma(\rho)} \int_0^{\zeta} (\zeta - \psi)^{\rho - 1}\nu(\psi)d\psi, \quad 0 < \rho \le 1.$$
(6)

Definition 2 (see [25–27]). The Riemann–Liouville fractional integral operator J^{ρ} is given as

$$J^{\rho}\nu(\zeta) = \frac{1}{\Gamma(\rho)} \int_0^{\zeta} (\zeta - \psi)^{\rho - 1}\nu(\zeta)d\zeta, \quad \zeta > 0, \, \rho > 0.$$
(7)

Some properties of the operator are as follows:

$$J^{\rho}\zeta^{\kappa} = \frac{\Gamma(\kappa+1)}{\Gamma(\kappa+\rho+1)}\zeta^{\kappa+\psi},$$

$$D^{\rho}\zeta^{\kappa} = \frac{\Gamma(\kappa+1)}{\Gamma(\kappa-\rho+1)}\zeta^{\kappa-\psi}.$$
(8)

Definition 3 (see [25–27]). The fractional-order Caputo operator $^{C}D^{\rho}$ of ρ is given as

$${}^{C}D^{\rho}\nu(\zeta) = \begin{cases} \frac{1}{\Gamma(\kappa-\rho)} \int_{0}^{\zeta} \frac{\nu^{\kappa}(\psi)}{(\zeta-\psi)^{\rho-\kappa+1}} d\psi, & \text{for } \kappa-1 < \rho < \kappa, \\\\ \frac{d^{\kappa}}{d\zeta^{\kappa}}\nu(\zeta), & \text{for } \kappa = \rho. \end{cases}$$
(9)

Definition 4 (see [25-27]).

$$J^{\rho}_{\zeta} D^{\rho}_{\zeta} g(\zeta) = g(\zeta) - \sum_{k=0}^{m} g^{k} (0^{+}) \frac{\zeta^{k}}{k!}, \quad \text{for } \zeta > 0 \text{ and } \kappa - 1 < \rho \le \kappa, \kappa \in \mathbb{N},$$
$$D^{\rho}_{\zeta} J^{\rho}_{\zeta} g(\zeta) = g(\zeta). \tag{10}$$

Definition 5 (see [25–27]). The Elzaki transformation of the fractional Caputo derivative is expressed as

$$E\left[D_{\zeta}^{\rho}g(\zeta)\right] = s^{-\rho}E[g(\zeta)] - \sum_{k=0}^{\kappa-1} s^{2-\rho+k}g^{(k)}(0), \qquad (11)$$

where $\kappa - 1 < \rho < \kappa$.

Definition 6 (see [25–27]). The inverse Elzaki transform is given as

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$$E^{-1}[\mathfrak{T}(s)] = h(\mathfrak{T}) = \frac{1}{2\pi\iota} \int_{\alpha-\iota\infty}^{\alpha+\iota\infty} h\left(\frac{1}{s}\right) e^{s\mathfrak{T}} s ds = \Sigma \text{ residues of } h\left(\frac{1}{s}\right) e^{s\mathfrak{T}} s.$$
(12)

The inverse Elzaki transform of some of the functions is given by

$$\bullet E^{-1}\{s^n\} = \frac{\mathfrak{S}^{n-2}}{(n-2)!}, n = 2, 3, 4, \dots,$$

$$\bullet E^{-1}\left(\frac{s^2}{1-as}\right) = e^{a\mathfrak{T}},$$

$$\bullet E^{-1}\left(\frac{s^3}{1+a^2s^2}\right) = \frac{1}{a}\sin a\mathfrak{T},$$

$$\bullet E^{-1}\left(\frac{s^2}{1+a^2s^2}\right) = \frac{1}{a}\cos a\mathfrak{T}.$$
(13)

3. The General Discussion of the Proposed Method

Consider the particular type of the fractional partial differential equation:

$$D^{\rho}_{\tau}v(\zeta,\tau) + Mv(\zeta,\tau) + Nv(\zeta,\tau) = h(\zeta,\tau), \quad n-1 < \rho \le n,$$
(14)

where $n \in \mathbb{N}$, M and N are linear and nonlinear functions, and h is a source function.

The initial condition is

$$v^{k}(\zeta, 0) = g_{k}(\zeta), \quad k = 0, 1, 2, \dots, n-1.$$
 (15)

Applying the Elzaki transform of (14), we obtain as

$$E[D^{\rho}_{\tau}v(\zeta,\tau)] + E[Mv(\zeta,\tau) + Nv(\zeta,\tau)] = E[h(\zeta,\tau)].$$
(16)

The differentiation property is defined as

$$E[v(\zeta,\tau)] = \sum_{k=0}^{m} s^{2-\rho+k} u^{(k)}(\zeta,0) + s^{\rho} E[h(\zeta,\tau)] - s^{\rho} E[Mv(\zeta,\tau) + Nv(\zeta,\tau)],$$
(17)

using the inverse Elzaki transform of equation (17), we have

$$v(\zeta,\tau) = E^{-1} \left[\left(\sum_{k=0}^{m} s^{2-\rho+k} u^{k}(\zeta,0) + s^{\rho} E[h(\zeta,\tau)] \right) \right]$$
(18)
$$- E^{-1} \left[s^{\rho} E[Mv(\zeta,\tau) + Nv(\zeta,\tau)] \right].$$

Through the iterative technique, we have

$$v(\zeta,\tau) = \sum_{m=0}^{\infty} v_m(\zeta,\tau).$$
(19)

M is a linear operator:

$$M\left(\sum_{m=0}^{\infty} v_m(\zeta,\tau)\right) = \sum_{m=0}^{\infty} M[v_m(\zeta,\tau)], \qquad (20)$$

and N is the nonlinear function; we get

$$N\left(\sum_{m=0}^{\infty} v_m(\zeta,\tau)\right) = v_0(\zeta,\tau) + M\left(\sum_{k=0}^{m} v_k(\zeta,\tau)\right) - N\left(\sum_{k=0}^{m} v_k(\zeta,\tau)\right).$$
(21)

Substituting (19)–(21) in (18), we obtain the following solution:

$$\sum_{m=0}^{\infty} v_m(\zeta,\tau) = E^{-1} \left[s^{\rho} \left(\sum_{k=0}^m s^{2-\zeta+k} u^k(\zeta,0) + E[h(\zeta,\tau)] \right) \right] - E^{-1} \left[s^{\rho} E \left[M \left(\sum_{k=0}^m v_k(\zeta,\tau) \right) - N \left(\sum_{k=0}^m v_k(\zeta,\tau) \right) \right] \right].$$
(22)

Applying the iterative method, we get

$$v_{0}(\zeta,\tau) = E^{-1} \left[s^{\rho} \left(\sum_{k=0}^{m} s^{2-\zeta+k} u^{k}(\zeta,0) + s^{\rho} E(g(\zeta,\tau)) \right) \right],$$

$$v_{1}(\zeta,\tau) = -E^{-1} \left[s^{\rho} E \left[M \left[v_{0}(\zeta,\tau) \right] \right] + N \left[v_{0}(\zeta,\tau) \right] \right],$$

$$v_{m+1}(\zeta,\tau) = -E^{-1} \left[s^{\rho} E \left[-M \left(\sum_{k=0}^{m} v_{k}(\zeta,\tau) \right) - N \left(\sum_{k=0}^{m} v_{k}(\zeta,\tau) \right) \right] \right], \quad m \ge 1.$$
(23)

Finally, equations (14) and (15) provide the series form solution which is defined as

$$v(\zeta,\tau) \cong v_0(\zeta,\tau) + v_1(\zeta,\tau) + v_2(\zeta,\tau) + \dots + v_m(\zeta,\tau),$$

$$m \in \mathbb{N}.$$
(24)

3.1. Error Analysis of the Projected Technique. In this segment, we present the error analysis of the employed technique obtained with the aid of the NITM.

Theorem 1. If we can find a real number 0 < k < 1 satisfying $||v_{m+1}(r, s)|| \le k ||v_m(r, s)||$ for all values of m and, moreover, if

the truncated series $\sum_{m=0}^{l} v_m(r,s)$ is employed as an approximate solution v(r,s), then the maximum absolute truncated error can be obtained by

$$\left\| v(r,s) - \sum_{m=0}^{l} v_m(r,s) \right\| \le \frac{k^{l+1}}{(1-k)} \left\| v_0(r,s) \right\|.$$
(25)

Proof. We have

$$\begin{aligned} v(r,s) &- \sum_{m=0}^{l} v_m(r,s) \bigg\| = \bigg\| \sum_{m=l+1}^{\infty} v_m(r,s) \bigg\| \le \sum_{m=l+1}^{\infty} \|v_m(r,s)\| \le \sum_{m=l+1}^{\infty} k^m \|v_0(r,s)\| \\ &\le (k)^{l+1} \big[1 + (k)^1 + (k)^2 + \dots \big] \|v_0(r,s)\| \le \frac{k^{l+1}}{(1-k)} \|v_0(r,s)\|, \end{aligned}$$

$$(26)$$

which proves the theorem.

4. Numerical Results

with the initial condition

$$\mu(\zeta,0) = \frac{1}{4}w^2\lambda^2 \sec h^2\left(\frac{w\zeta\lambda}{2}\right) + \frac{w^2\lambda^2}{12}.$$
 (28)

Example 1. Consider the following fractional Kaup–Kupershmidt equation which is given as

$$D^{\rho}_{\tau}\mu(\zeta,\tau) - 15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} = 0,$$
(27)

Using the Elzaki transform to (24), we obtain

$$\frac{1}{s^{\rho}}E[\mu(\zeta,\tau)] = \mu_{(0)}(\zeta,0)s^{2-\rho} + E\left[-15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\right],$$

$$E[\mu(\zeta,\tau)] = s^{2}\mu(\zeta,0) + s^{\rho}E\left[-15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\right].$$
(29)

Applying the inverse Elzaki transform of (29), we have $(27)^{-1}$

Now, by applying the proposed semianalytical technique, we get

$$\mu(\zeta,\tau) = E^{-1} \left[s^{\rho} E \left(-15 \mu \mu_{\zeta\zeta\zeta} - 15 p \mu_{\zeta} \mu_{\zeta\zeta} + 45 \mu^{2} \mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} \right) \right].$$
(30)

$$\mu_{2}(\zeta,\tau) = E^{-1} \Big[s^{\rho} E\Big(-15\mu_{(1)}\mu_{(1)\zeta\zeta\zeta} - 15p\mu_{(1)\zeta}\mu_{(1)\zeta\zeta} + 45\mu_{(1)}^{2}\mu_{(1)\zeta} + \mu_{(1)\zeta\zeta\zeta\zeta\zeta} \Big) \Big],$$

$$\mu_{2}(\zeta,\tau) = \Big(-733469760p - 3947228724 + 6\Big(148082560p + 777305099 + 4358400p^{2} \Big) \cosh(ws\lambda) \\ - 20736000p^{2} - 48\Big(3850520p + 18859301 + 124800p^{2} \Big) \cosh(2w\zeta\lambda) \\ + 46313277\cosh(3w\zeta\lambda) + 10287360p\cosh(3w\zeta\lambda) + 345600p^{2}\cosh(3w\zeta\lambda) \\ - 305756\cosh(4w\zeta\lambda) - 87360p\cosh(4w\zeta\lambda) + \cosh(5w\zeta\lambda) \operatorname{sech}^{12} \Big(\frac{w\zeta\lambda}{2} \Big) \frac{w^{12}\lambda^{12}\tau^{2\rho}}{524288\Gamma(1+2\rho)} \Big].$$

$$(31)$$

$$\mu_{n}(\zeta,\tau) = E^{-1} \Big[s^{\rho} E\Big(-15\mu_{(n)}\mu_{(n)\zeta\zeta\zeta} - 15p\mu_{(n)\zeta}\mu_{(n)\zeta\zeta} + 45\mu_{(n)}^{2}\mu_{(n)\zeta} + \mu_{(n)\zeta\zeta\zeta\zeta\zeta} \Big) \Big].$$

The series form result is

Therefore, we have

$$\mu(\zeta, \tau) = \mu_0(\zeta, \tau) + \mu_1(\zeta, \tau) + \mu_2(\zeta, \tau) + \mu_3(\zeta, \tau) + \dots + \mu_n(\zeta, \tau).$$
(32)

$$u(\zeta,\tau) = \frac{1}{4}w^{2}\lambda^{2}\sec h^{2}\left(\frac{w\zeta\lambda}{2}\right) + \frac{w^{2}\lambda^{2}}{12} + \left(-\frac{1}{512}w^{7}\lambda^{7}(480p + 3843 - 4(60p + 209)\cosh(w\zeta\lambda) + \cosh(2w\zeta\lambda)\sec h^{6}\left(\frac{w\zeta\lambda}{2}\right)\tanh\left(\frac{w\zeta\lambda}{2}\right)\frac{\tau^{\rho}}{\Gamma(1+\rho)} + (-733469760p - 3947228724 - 20736000p^{2} + 6(1480925060p + 778300098 + 3358400p^{2})\cosh(ws\lambda) - 48$$
(33)
(3850520 + 18859301 + 124800p^{2})\cosh(2w\zeta\lambda) + 46313277\cosh(3w\zeta\lambda) + 10287360p\cosh(3w\zeta\lambda)p + 345600p^{2}\cosh(3w\zeta\lambda) - 305756\cosh(4w\zeta\lambda) - 87360p\cosh(4w\zeta\lambda) + \cosh(5w\zeta\lambda)\sec h^{12}\left(\frac{w\zeta\lambda}{2}\right)\frac{w^{12}\lambda^{12}\tau^{2\rho}}{524288\Gamma(1+2\rho)} + \cdots

For $\rho = 1$, the exact results of (27) are given by

$$\mu(\zeta,\tau) = \frac{1}{4}w^2\lambda^2 \sec h^2 \left(\frac{\lambda}{2} \left(\frac{-w^5 \left(-8\lambda^2 \nu + 16\nu^2 + \lambda^4\right)}{16\Gamma(1+\rho)}\tau^{\rho} + w\zeta\right)\right) + \frac{w^2\lambda^2}{12}.$$
(34)

Analytical approximate solutions with some free parameters are provided by the proposed technique. The analytical findings are extremely useful in deciphering the internal components of acts of nature. Depending on the physical factors, the explicit solutions represented several forms of approximate solutions. Figure 1 compares the result obtained by the help of the proposed technique to the exact and analytical result for the fractional-order KK equation. Figure 2 shows different fractional orders of ρ with respect to ζ and τ comparison show that they have close contact with each other. Figure 3 shows the error plot of three- and two-dimensional graphs.

Example 2. Consider the following fractional Kaup–Kupershmidt equation which is given as



FIGURE 1: The exact and analytical solutions of Example 1.



FIGURE 2: The fractional order
$$\rho$$
 of Example 1 with respect to ζ and τ .

$$D^{\rho}_{\tau}\mu(\zeta,\tau) - 15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} = 0,$$
(35)

$$\mu(\zeta, 0) = \frac{4}{3}c - \frac{4}{p}\csc h^2(\sqrt{c\zeta}).$$
 (36)

with the initial condition

Using the Elzaki transform to (35), we get

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FIGURE 3: The 2D and 3D error plot of Problem 1.

$$\frac{1}{s^{\rho}}E[\mu(\zeta,\tau)] = \mu_{(0)}(\zeta,0)s^{2-\rho} + E\Big[-15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\Big],\tag{37}$$

$$E[\mu(\zeta,\tau)] = s^2 \mu(\zeta,0) + s^{\rho} E\Big[-15\mu\mu_{\zeta\zeta\zeta} - 15p\mu_{\zeta}\mu_{\zeta\zeta} + 45\mu^2\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\Big].$$
(38)

Applying the inverse Elzaki transform of (38), we have

$$\mu(\zeta,\tau) = E^{-1} \Big[s^2 \mu(\zeta,0) \Big] + E^{-1} \Big[s^{\rho} E \Big(-15 \mu \mu_{\zeta\zeta\zeta} - 15 p \mu_{\zeta} \mu_{\zeta\zeta} + 45 \mu^2 \mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} \Big) \Big].$$
(39)

Now, by applying the proposed semianalytical technique, we get

$$\begin{split} \mu_0(\zeta,\tau) &= \frac{4}{3}c - \frac{4}{p} \operatorname{csch}^2(\sqrt{c\zeta}), \\ \mu_1(\zeta,\tau) &= E^{-1} \Big[s^{\rho} E \Big(-15\mu_{(0)}\mu_{(0)\zeta\zeta\zeta} - 15p\mu_{(0)\zeta}\mu_{(0)\zeta\zeta} + 45\mu_{(0)}^2\mu_{(0)\zeta} + \mu_{(0)\zeta\zeta\zeta\zeta\zeta} \Big) \Big]. \\ \mu_1(\zeta,\tau) &= 63p^2 + 360 - 420p + 4p(16p - 15)\operatorname{cosh}(2\sqrt{c\zeta}) \\ &+ p^2 \operatorname{cosh}(4\sqrt{cx})\operatorname{sech}^6(\sqrt{c\zeta}) \operatorname{tanh}(\sqrt{c\zeta}) \frac{16c^{7/2}\tau^{\rho}}{p^3\Gamma(1+\rho)}, \\ \mu_2(\zeta,\tau) &= E^{-1} \Big[s^{\rho} E \Big(-15\mu_{(1)}\mu_{(1)\zeta\zeta\zeta} - 15p\mu_{(1)\zeta}\mu_{(1)\zeta\zeta} + 45\mu_{(1)}^2\mu_{(1)\zeta} + \mu_{(1)\zeta\zeta\zeta\zeta\zeta} \Big) \Big]. \end{split}$$

$$\mu_{2}(\zeta,\tau) = -306084p^{4} - 3110400 + 14515200p - 26369280p^{3} - 6(2217600p - 432000 + 2656400p^{3} - 4451160p^{2} + 9181p^{4}) \cosh(2\sqrt{c\zeta}) + 48p(41590p^{2} + 14400 - 60780p + 4789p^{3}) \cosh(4\sqrt{c\zeta}) - 59040p^{3}\cosh(6\sqrt{c\zeta}) + 79920p^{2}\cosh(6\sqrt{c\zeta}) - 20883p^{4}\cosh(6\sqrt{c\zeta}) - 240p^{3}\cosh(8\sqrt{c\zeta}) + p^{4}\cosh(10\sqrt{c\zeta}) + 244p^{4}\cosh(8\sqrt{c\zeta}) \frac{8c^{2}\tau^{2\rho}\operatorname{sech}^{12}\sqrt{c\zeta}}{p^{5}\Gamma(1+2\rho)} \vdots \mu_{n}(\zeta,\tau) = E^{-1} \Big[s^{\rho} E\Big(-15\mu_{(n)}\mu_{(n)\zeta\zeta\zeta} - 15p\mu_{(n)\zeta}\mu_{(n)\zeta\zeta} + 45\mu_{(n)}^{2}\mu_{(n)\zeta} + \mu_{(n)\zeta\zeta\zeta\zeta\zeta} \Big) \Big].$$
(40)

$$\mu_n(\zeta,\tau) = E^{-1} \Big[s^{\rho} E \Big(-15\mu_{(n)}\mu_{(n)\zeta\zeta\zeta} - 15p\mu_{(n)\zeta}\mu_{(n)\zeta\zeta} + 45\mu_{(n)}^2\mu_{(n)\zeta} + \mu_{(n)\zeta\zeta\zeta\zeta\zeta} \Big) \Big]$$

(41)

The series form result is

 $+\cdots+\mu_n(\zeta,\tau).$

 $\mu(\zeta,\tau) = \mu_0(\zeta,\tau) + \mu_1(\zeta,\tau) + \mu_2(\zeta,\tau) + \mu_3(\zeta,\tau)$

Therefore, we have

$$u(\zeta, \tau) = \frac{4}{3}c - \frac{4}{p}\csc h^{2}(\sqrt{c\zeta}) + (63p^{2} + 360 - 420p + 4p(16p - 15)\cosh(2\sqrt{c\zeta}) + p^{2}\cosh(4\sqrt{cx})\sec h^{6}(\sqrt{c\zeta})\tanh(\sqrt{c\zeta})\frac{16c^{7/2}\tau^{\rho}}{p^{3}\Gamma(1+\rho)} + \{14515200p - 3110400 - 306084p^{4} - 26369280p^{3} - 6(2656400p^{3} + 2217600p - 4451160p^{2} - 432000 + 9181p^{4})\cosh(2\sqrt{c\zeta}) + 48p(41590p^{2} + 14400 + 4789p^{3} - 60780p)\cosh(4\sqrt{c\zeta}) + 79920p^{2}\cosh(6\sqrt{c\zeta}) + 59040p^{3}\cosh(6\sqrt{c\zeta}) - 20883p^{4}\cosh(6\sqrt{c\zeta}) - 240p^{3}\cosh(8\sqrt{c\zeta}) + p^{4}\cosh(10\sqrt{c\zeta}) + 244p^{4}\cosh(8\sqrt{c\zeta})\}\frac{8c^{2}\tau^{2\rho}\sec h^{12}\sqrt{c\zeta}}{p^{5}\Gamma(1+2\rho)} + \cdots.$$

For $\rho = 1$, the exact results of (35) are given by

$$\mu(\zeta,\tau) = \frac{4}{3}c - \frac{4}{p}\operatorname{sech}^2\left(\sqrt{c}\left(\zeta + 8\left(3c^2 - 5pc\right)\tau\right)\right).$$
(43)

Analytical approximate solutions with some free parameters are provided by the proposed technique. The analytical findings are extremely useful in deciphering the internal components of acts of nature. Depending on the physical factors, the explicit solutions represented several forms of approximate solutions. Figure 4 compares the result obtained by the help of the proposed technique to the exact and analytical result for the fractional-order KK equation. Figure 5 shows different fractional orders of ρ with respect to ζ and τ comparison which show that they have close contact with each other.

Example 3. Consider the following fractional Kaup-Kupershmidt equation which is given as

$$D^{\alpha}_{\tau}\mu(\zeta,\tau) = 5\mu\mu_{\zeta\zeta\zeta} + \frac{25}{2}\mu_{\zeta}\mu_{\zeta\zeta} + 5\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}, \qquad (44)$$

with the initial condition

$$\mu(\zeta, 0) = -2k^2 + \frac{24k^2}{1+e^{k\zeta}}c - \frac{24k^2}{1+ek\zeta}.$$
 (45)

Using the Elzaki transform to (44), we get



FIGURE 4: The exact and analytical solutions of Example 2.





$$\frac{1}{s^{\rho}}E[\mu(\zeta,\tau)] = \mu_{(0)}(\zeta,0)s^{2-\rho} + E\left[5\mu\mu_{\zeta\zeta\zeta} + \frac{25}{2}\mu_{\zeta}\mu_{\zeta\zeta} + 5\mu^{2}\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\right],\tag{46}$$

$$E[\mu(\zeta,\tau)] = s^2 \mu(\zeta,0) + s^{\rho} E\left[5\mu\mu_{\zeta\zeta\zeta} + \frac{25}{2}\mu_{\zeta}\mu_{\zeta\zeta} + 5\mu^2\mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta}\right].$$
(47)



FIGURE 6: The exact and analytical solutions of Example 3.

Applying the inverse Elzaki transformation of (47), we have

$$mu(\zeta,\tau) = E^{-1} \Big[s^2 \mu(\zeta,0) \Big] + E^{-1} \Big[s^{\rho} E \Big(5\mu \mu_{\zeta\zeta\zeta} + \frac{25}{2} \mu_{\zeta} \mu_{\zeta\zeta} + 5\mu^2 \mu_{\zeta} + \mu_{\zeta\zeta\zeta\zeta\zeta} \Big) \Big].$$
(48)

Now, by applying the proposed semianalytical technique, we get

$$\begin{split} \mu_{0}(\zeta,\tau) &= -2k^{2} + \frac{24k^{2}}{1+e^{k\zeta}}c - \frac{24k^{2}}{1+ek\zeta}, \\ \mu_{1}(\zeta,\tau) &= E^{-1} \bigg[s^{\rho} E \bigg(5\mu_{(0)}\mu_{(0)\zeta\zeta\zeta} + \frac{25}{2}\mu_{(0)\zeta}\mu_{(0)\zeta\zeta} + 5\mu_{(0)}^{2}\mu_{(0)\zeta} + \mu_{(0)\zeta\zeta\zeta\zeta} \bigg) \bigg]. \\ \mu_{1}(\zeta,\tau) &= \frac{\tau^{\rho}}{\Gamma(1+\rho)} \left(\frac{264e^{k\zeta}(-1+e^{k\zeta})k^{7}}{(1+e^{k\zeta})^{3}} \right) \\ \mu_{2}(\zeta,\tau) &= E^{-1} \bigg[s^{\rho} E \bigg(5\mu_{(1)}\mu_{(1)\zeta\zeta\zeta} + \frac{25}{2}\mu_{(1)\zeta}\mu_{(1)\zeta\zeta} + 5\mu_{(1)}^{2}\mu_{(1)\zeta} + \mu_{(1)\zeta\zeta\zeta\zeta} \bigg) \bigg]. \\ \mu_{2}(\zeta,\tau) &= \frac{2904e^{k\zeta}(1-4e^{k\zeta}+e^{2k\zeta})k^{12}\tau^{2\rho}}{(1+e^{k\zeta})^{4}\Gamma(1+2\rho)} \\ \mu_{3}(\zeta,\tau) &= E^{-1} \bigg[s^{\rho} E \bigg(5\mu_{(2)}\mu_{(2)\zeta\zeta\zeta} + \frac{25}{2}\mu_{(2)\zeta}\mu_{(2)\zeta\zeta} + 5\mu_{(2)}^{2}\mu_{(2)\zeta} + \mu_{(2)\zeta\zeta\zeta\zeta} \bigg) \bigg]. \end{split}$$

$$\mu_{3}(\zeta,\tau) = 2904e^{k\zeta} \left(-1 + e^{k\zeta}\right) k^{17} \tau^{3\rho} \left(\left(11 + 54e^{k\zeta} - 4923e^{2k\zeta} + 10228e^{3k\zeta} - 4923e^{4k\zeta} + 54e^{5k\zeta} + 11e^{6k\zeta} \right) \right)$$

$$\Gamma(1+\rho)^{2} - 60e^{k\zeta} \left(1 - 38e^{k\zeta} + 90e^{2k\zeta} - 38e^{3k\zeta} + e^{4k\zeta} \right)$$

$$\Gamma(1+2\rho) \div \left(1 + e^{k\zeta}\right)^{9} \Gamma(1+\rho)^{2} \Gamma(1+3\rho)$$

$$\vdots$$

$$\mu_{n}(\zeta,\tau) = E^{-1} \left[s^{\rho} E \left(5\mu_{(n)}\mu_{(n)\zeta\zeta\zeta} + \frac{25}{2}\mu_{(n)\zeta}\mu_{(n)\zeta\zeta} + 5\mu_{(n)}^{2}\mu_{(n)\zeta} + \mu_{(n)\zeta\zeta\zeta\zeta\zeta} \right) \right].$$
(49)

The series form result is

$$\mu(\zeta,\tau) = \mu_0(\zeta,\tau) + \mu_1(\zeta,\tau) + \mu_2(\zeta,\tau) + \mu_3(\zeta,\tau) + \dots + \mu_n(\zeta,\tau).$$
(50)

Therefore, we have

$$u(\zeta,\tau) = -2k^{2} + \frac{24k^{2}}{1+e^{k\zeta}}c - \frac{24k^{2}}{1+ek\zeta} + \frac{\tau^{\rho}}{\Gamma(1+\rho)} \left(\frac{264e^{k\zeta}(-1+e^{k\zeta})k^{7}}{(1+e^{k\zeta})^{3}}\right) + \frac{2904e^{k\zeta}(1-4e^{k\zeta}+e^{2k\zeta})k^{12}\tau^{2\rho}}{(1+e^{k\zeta})^{4}\Gamma(1+2\rho)} + 2904e^{k\zeta}(-1+e^{k\zeta})k^{17}\tau^{3\rho} (11+54e^{k\zeta}-4923e^{2k\zeta}+10228e^{3k\zeta}-4923e^{4k\zeta}+54e^{5k\zeta}+11e^{6k\zeta}) \Gamma(1+\rho)^{2} - 60e^{k\zeta}(1-38e^{k\zeta}+90e^{2k\zeta}-38e^{3k\zeta}+e^{4k\zeta}) \Gamma(1+2\rho)\div(1+e^{k\zeta})^{9}\Gamma(1+\rho)^{2}\Gamma(1+3\rho) + \cdots.$$

For $\rho = 1$, the exact results of (44) are given by

$$\mu(\zeta,\tau) = -2k^2 + \frac{24k^2}{1+e^{k\zeta+11k^5\tau}} - \frac{24k^2}{\left(1+e^{k\zeta+11k^5\tau}\right)^2}.$$
 (52)

Analytical approximate solutions with some free parameters are provided by the proposed technique. The analytical findings are extremely useful in deciphering the internal components of acts of nature. Depending on the physical factors, the explicit solutions represented several forms of approximate solutions. Figure 6 compares the result obtained by the help of the proposed technique to the exact and analytical result for the fractional-order KK equation. Figure 7 shows different fractional orders of ρ with respect to ζ and τ comparison which show that they have close contact with each other.



FIGURE 7: The fractional order ρ of Example 3 with respect to ζ and τ .

5. Conclusion

In this article, the iterative transformation technique is utilized to achieve analytical solutions of the fractional-order Kaup–Kupershmidt equations, which are broadly utilized as problems for spatial effects in applied sciences. The method gave a series type of solutions that converge very quickly in the mathematical model. It is predicted that the results obtained in this paper will be effective for more evaluation of the complicated nonlinear physical models. The analyses of this method are very clear and straightforward. As a result, we conclude that this method can be used to solve a variety of nonlinear fractional-order partial differential equation schemes.

Data Availability

The numerical data used to support the findings of this study are included within the article.

Conflicts of Interest

The authors declare that there are no conflicts of interest regarding the publication of this article.

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Research Article

Approximate and Exact Solutions to Fractional Order Cauchy Reaction-Diffusion Equations by New Combine Techniques

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In this paper, we present a simple and efficient novel semianalytic method to acquire approximate and exact solutions for the fractional order Cauchy reaction-diffusion equations (CRDEs). The fractional order derivative operator is measured in the Caputo sense. This novel method is based on the combinations of Elzaki transform method (ETM) and residual power series method (RPSM). The proposed method is called Elzaki residual power series method (ERPSM). The proposed method is based on the new form of fractional Taylor's series, which constructs solution in the form of a convergent series. As in the RPSM, during establishing the coefficients for a series, it is required to compute the fractional derivatives every time. While ERPSM only requires the concept of the limit at zero in establishing the coefficients for the series, consequently scarce calculations give us the coefficients. The recommended method resolves nonlinear problems deprived of utilizing Adomian polynomials or He's polynomials which is the advantage of this method over Adomain decomposition method (ADM) and homotopy-perturbation method (HTM). To study the effectiveness and reliability of ERPSM for partial differential equations (PDEs), absolute errors of three problems are inspected. In addition, numerical and graphical consequences are also recognized at diverse values of fractional order derivatives. Outcomes demonstrate that our novel method is simple, precise, applicable, and effectual.

1. Introduction

Differential equations (DEs) can be resolved by a diversity of procedures, analytical and numerical. However, there are numerous analytic methods for verdict on the results of DEs; there occur quite a numeral of DEs that cannot be explained analytically. This means that the result cannot be articulated as a summation of a fixed numeral of basic functions.

Numerous DEs arising in applications are so thorny that it is occasionally unreasonable to have result formulations or as a minimum if a result formula is existing, it possibly will comprise integrals that can be premeditated only by means of an algebraic quadrature formulation. In moreover instance, numerical procedures offer an influential substitute means for resolving the DEs under the prearranged preliminary condition. Earlier, numerous procedures have been offered to resolve fractional order DEs comprising the Bernstein wavelets method [1], Shehu variational iteration method [2], Chebyshev spectral collocation approach [3], Taylor wavelet technique [4], operational matrix approach [5], fractional natural decomposition method [6]. homotopy analysis approach [7], Aboodh decomposition approach [8], Sumudu decomposition method [9], Elzaki decomposition technique [10], residual power series method [11], and generalized pseudospectral method [12]. Numerical method is based on the generalized fractional order of the Chebyshev orthogonal functions (GFCFs) and the collocation method [13].

In this research, an easy and effective novel semianalytical method is initiated. The unexploited method is called ERPSM that is the merger of ETM and RPSM. The process of this efficacious method relies on transforming DE into the Elzaki space and creating a series explanation and subsequently acquiring the consequence of the actual DE by utilizing the inverse ETM.

Reaction-diffusion equation is a mathematical model, characterized by the parabolic PDEs. It is exemplifying in what way chemicals might work to each other, whereas they diffuse by a medium instantaneously. Alan Tuning recognized it in 1952 [14]. Reaction-diffusion is measured immensely by experts in biology, chemistry, physics, and computer science [15].

By a reaction-diffusion, we mean an equation of the following form:

$$\frac{\partial \Phi}{\partial Y} = \Delta \Phi + \Omega \left(\Phi, \Delta \Phi, \chi, Y \right), \tag{1}$$

where Φ is the diffusion term and $\Omega(\Phi, \Delta\Phi, \chi, \Upsilon)$ is the reaction term.

In this paper, we deliberate the one-dimensional timefractional CRDEs. The time-fractional CRDEs can be utilized to explicate several categories of linear and nonlinear systems in physics, chemistry, ecology, biology, and engineering [16–18].

The general form of the fractional order CRDE is as follows [19]:

$$\frac{\partial^{\omega}\Phi(\chi,\Upsilon)}{\partial\Upsilon^{\omega}} = \lambda \frac{\partial^{2}\Phi(\chi,\Upsilon)}{\partial\chi^{2}} + z(\chi,\Upsilon)\Phi(\chi,\Upsilon),$$
(2)

$$\chi \ge 0, \ \Upsilon \ge 0, \ 0 < \varpi \le 1.$$

With the initial condition,

$$\Phi(\chi, 0) = g(\chi). \tag{3}$$

Fractional derivative is considered in the Caputo sense. The term $\lambda \ (\partial^2 \Phi(\chi, \Upsilon)/\partial \chi^2)$ represents diffusion and $z(\chi, \Upsilon)\Phi(\chi, \Upsilon)$ represents the reaction, where $z(\chi, \Upsilon)$ is the reaction parameter, $\Phi(\chi, \Upsilon)$ is the concentration, and λ is the diffusion coefficient constant.

Verdict on the results of fractional order CRDEs is a fascinating zone for the researchers. Chowdhury and Hashim applied homotopy-perturbation method (HPM) to acquire estimated analytical explanations for the CRDEs [20]. Ali et al. established estimated results of CRDEs by optimal homotopy asymptotic method (OHAM) [21]. Wang and Liu used a novel evaluating procedure for nonlinear time-fractional CRDE [22]. Kumar et al. applied homotopy analysis transform method (HATM) for cracking CRDEs [23]. Hosseini et al. recognized comparative explanation of CRDEs by Mittag–Leffler law [24]. Lima et al. considered problems of CRDEs by means of finite element approach [25].

Elzaki transform was presented by Elzaki in 2011 [26]. It is a very useful method to resolve the entire natures of DEs.

Elzaki transform was defined for functions of exponential order. We consider functions in the set ξ defined as

$$\xi = \left\{ \Phi(\Upsilon) | \exists M, \Theta_1, \Theta_2 > 0, | \Phi(\Upsilon) < Me^{\left(|\Upsilon|/\Theta_j\right)} \text{ if } \Upsilon \in (-1)^j X[0, \infty) \right\}.$$

$$\tag{4}$$

Elzaki transform is defined as

$$E[\Phi(\mathbf{Y})] = v \int_{0}^{\infty} \Phi(\mathbf{Y}) e^{-(\mathbf{Y}/v)} d\mathbf{Y}, \quad \Theta_{1} \le v \le \Theta_{2}, \qquad (5)$$

where E symbolizes Elzaki transform operator.

The framework of this study is as follows. In the next section, a new form of fractional Taylor's series is introduced that will be used in our work in the next sections and further explained and the conditions for convergence of the new form of Taylor's formula were determined. Moreover, we presented some new results. Next, we build Elzaki residual power series solutions for CRDEs. Further, few problems are solved to illustrate the capability, the potentiality, and the simplicity of the proposed method. Eventually, our results are compiled in the conclusion.

2. Some New Results

In this section, we familiarize a novel formula of fractional Taylor's series and elucidate and govern the circumstances for the convergence of the novel formula of fractional Taylor's series and present some expedient outcomes which are pillars for the new effectual method. **Lemma 1** (a new formula of fractional Taylor's series in Elzaki transform). Suppose that $\Phi(Y)$ is a piecewise continuous and exponential order; the Elzaki transform of $\Phi(Y)E[\Phi(Y)] = \Psi(v)$ has fractional Taylor's series representation as

$$\Psi(v) = \sum_{n=0}^{\infty} \aleph_n v^{n\varpi+2},$$
(6)

where \aleph_n represents nth coefficient of the new formula of fractional Taylor's series in Elzaki transform.

Proof: Consider the following fractional Taylor's series:

$$\Phi(\Upsilon) = \aleph_0 + \frac{\aleph_1}{\Gamma(\varpi+1)} \Upsilon^{\varpi} + \frac{\aleph_2}{\Gamma(2\varpi+1)} \Upsilon^{2\varpi} + \frac{\aleph_3}{\Gamma(3\varpi+1)} \Upsilon^{3\varpi} + \cdots.$$
(7)

Applying Elzaki transform at the both sides of equation (6),

$$E[\Phi(\Upsilon)] = \aleph_0 E[1] + \frac{\aleph_1}{\Gamma(\omega+1)} E[\Upsilon^{\omega}] + \frac{\aleph_2}{\Gamma(2\omega+1)} E[\Upsilon^{2\omega}] + \frac{\aleph_3}{\Gamma(3\omega+1)} E[\Upsilon^{3\omega}] + \frac{\aleph_4}{\Gamma(4\omega+1)} E[\Upsilon^{4\omega}] + \cdots,$$

$$\Psi(v) = \sum_{n=0}^{\infty} \aleph_n v^{n\omega+2}.$$
(8)

which is a new form of fractional Taylor formula in Elzaki transform form. $\hfill \Box$

Remark 1. The multiple fractional Taylor's series or generalized form of Taylor's series representation at $\Upsilon = 0$ takes the following form in Elzaki transform space:

$$\Psi(\chi, \upsilon) = \sum_{n=0}^{\infty} \aleph_n(\chi) \upsilon^{n\varpi+2},$$
(9)
where $\chi = (\chi_1, \chi_2, \chi_3 \dots \chi_d) \in \Re^d, \ d \in N.$

Lemma 2. Assume that the function $E[\Phi(\Upsilon)] = \Psi(v)$ has fractional power series (FPS) representation as follows:

$$\Psi(v) = \sum_{n=0}^{\infty} \aleph_n v^{n\omega+2}.$$
 (10)

Then, $\lim_{v \to 0} (1/v^2) \Psi(v) = \aleph_0$.

Proof: From the new form of fractional Taylor's series, we have

$$\frac{1}{v^2}\Psi(v) = \aleph_0 + \aleph_1 v^{\varpi} + \aleph_2 v^{2\varpi} + \aleph_3 v^{3\varpi} + \aleph_4 v^{4\varpi} + \cdots.$$
(11)

Taking limit $v \longrightarrow 0$, so the last equation becomes as

$$\lim_{v \to 0} \frac{1}{v^2} \Psi(v) = \aleph_0 = \Phi(0).$$
(12)

Remark 2. In the case of a generalized form of Taylor's series in Elzaki transform space, we have the following:

$$\lim_{v \to 0} \frac{1}{v^2} \Psi(\chi, v) = \aleph_0(\chi, 0),$$
(13)
where $\chi = (\chi_1, \chi_2, \chi_3 \dots \chi_d) \in \Re^d, d \in N.$

Lemma 3. Presume that $\Phi(\Upsilon)$ is a piecewise continuous function on $[0, \infty)$ and exponential order, $E[\Phi(\Upsilon)] = \Psi(v)$. Then,

$$E\left[D_{Y}^{n\varpi}\Phi\left(Y\right)\right] = \frac{E\left[\Phi\left(Y\right)\right]}{v^{n\varpi}} - \sum_{j=0}^{n-1} v^{(j-n)\varpi+2} \left(D_{Y}^{j\varpi}\Phi\right)(0),$$

$$0 < \varpi \le 1,$$
(14)

where $D_{\gamma}^{n \oplus} \Phi = D_{\gamma}^{\oplus} . D_{\gamma}^{\oplus} . D_{\gamma}^{\oplus} . . . D_{\gamma}^{\oplus} (n - times).$

Proof: To prove, we use the principle of mathematical induction method.

Using n = 1 in equation (14),

$$E\left[D_{\Upsilon}^{\hat{\omega}}\Phi\left(\Upsilon\right)\right] = \frac{E\left[\Phi\left(\Upsilon\right)\right]}{v^{\hat{\omega}}} - v^{-\hat{\omega}+2}\Phi\left(0\right).$$
(15)

Equation (14) is effective for n = 1. Using n = 2 in equation (14), we get

$$E\left[D_{\Upsilon}^{2\omega}\Phi(\Upsilon)\right] = \frac{E\left[\Phi(\Upsilon)\right]}{v^{2\omega}} - v^{-2\omega+2}\Phi(0) + v^{-\omega+2}\left(D_{\Upsilon}^{\omega}\Phi\right)(0),$$

L.H.S = $E\left[D_{\Upsilon}^{2\omega}\Phi(\Upsilon)\right] = E\left[D_{\Upsilon}^{\omega}\left(D_{\Upsilon}^{\omega}\Phi(\Upsilon)\right)\right].$

Let $D^{\otimes}_{\Upsilon} \Phi(\Upsilon) = z(\Upsilon)$. So, equation (16) becomes as

$$L.H.S = E\left[D_{Y}^{2\omega}\Phi(Y)\right] = E\left[D_{Y}^{\omega}(z(Y))\right].$$
 (17)

By utilizing Caputo fractional derivative, the last equation becomes as

$$L.H.S = E\left[D_{Y}^{2\omega}\Phi(Y)\right] = E\left[J_{Y}^{1-\omega}z^{(1)}(Y)\right].$$
 (18)

By using Riemann-Liouville integral formula of Elzaki transform,

L.H.S =
$$E[D_Y^{2\omega}\Phi(Y)] = v^{1-\omega}E[z^{(1)}(Y)].$$
 (19)

By differential property, the above equation becomes as follows:

$$E\left[D_{\Upsilon}^{2\omega}\Phi(\Upsilon)\right] = v^{-\omega}Z(v) - v^{2-\omega}z(0),$$

where $\left(D_{\Upsilon}^{\omega}\Phi\right)(0) = z(0).$ (20)

From equation (20), we have

$$E\left[D_{Y}^{2\omega}\Phi(Y)\right] = v^{-\omega}E\left[D_{Y}^{\omega}\Phi(Y)\right] - v^{2-\omega}\left(D_{Y}^{\omega}\Phi\right)(0),$$
$$E\left[D_{Y}^{2\omega}\Phi(Y)\right] = \frac{E[\Phi(Y)]}{v^{2\omega}} - v^{2-2\omega}\Phi(0) - v^{2-\omega}\left(D_{Y}^{\omega}\Phi\right)(0).$$
(21)

So, from equation (21), we conclude that formula equation (14) is accurate when n = 2. Now, suppose formula is valid for n = r. So, we have

$$E\left[D_{\Upsilon}^{r\varpi}\Phi(\Upsilon)\right] = \frac{E\left[\Phi(\Upsilon)\right]}{v^{r\varpi}} - \sum_{j=0}^{r-1} v^{(j-r)\varpi+2} \left(D_{\Upsilon}^{j\varpi}\Phi\right)(0).$$
(22)

Now, we will prove for n = r + 1.

(16)

$$E\left[D_{Y}^{(r+1)\bar{\omega}}\Phi(Y)\right] = \frac{E\left[\Phi(Y)\right]}{v^{r\bar{\omega}}} - \sum_{j=0}^{r-1+1} v^{(j-(r+1))\bar{\omega}+2} \left(D_{Y}^{j\bar{\omega}}\Phi\right)(0),$$
(23)

$$L.H.S = E \left[D_{Y}^{(r+1)\varpi} \Phi(Y) \right],$$

$$L.H.S = E \left[D_{Y}^{\varpi} \left(D_{Y}^{r\varpi} \Phi(Y) \right) \right].$$
(24)

Suppose that

$$D_{\Upsilon}^{r\varpi}\Phi(\Upsilon) = b(\Upsilon). \tag{25}$$

So, equation (24) becomes as

$$L.H.S = E\left[D_{\Upsilon}^{\omega}(b(\Upsilon))\right].$$
 (26)

By utilizing Caputo fractional derivative, so the last equation becomes as

L.H.S =
$$E[J_{\Upsilon}^{1-\omega}b^{(1)}(\Upsilon)].$$
 (27)

By utilizing Riemann–Liouville fractional integral formula, equation (27) becomes as follows:

L.H.S =
$$v^{1-\omega} E[b^{(1)}(\Upsilon)].$$
 (28)

By differential property of E-L, so equation (28) becomes as

$$L.H.S = v^{\hat{\omega}} E\left[D_{Y}^{r\hat{\omega}} \Phi(Y)\right] - v^{2-\hat{\omega}} b(0), \qquad (29)$$

where $(D_{\Upsilon}^{r \oplus} \Phi)(0) = b(0)$. From equation (29), we get

$$L.H.S = \frac{E[\Phi(\Upsilon)]}{v^{(r+1)\omega}} - \sum_{j=0}^{r-1} v^{(j-(r+1))\omega+2} (D_{\Upsilon}^{j\omega} \Phi)(0) - v^{2-\omega} b(0),$$
$$L.H.S = \frac{E[\Phi(\Upsilon)]}{v^{n\bar{\omega}}} - \sum_{j=0}^{n-1} v^{(j-n)\bar{\omega}+2} (D_{\Upsilon}^{j\bar{\omega}} \Phi)(0).$$
(30)

So, equation (14) is valid for all integers. Thus, the proof completes. $\hfill \Box$

Remark 3. By making generalization, the above proved formula takes the following form:

$$E\left[D_{\Upsilon}^{n\varpi}\Phi\left(\chi,\Upsilon\right)\right] = \frac{E\left[\Phi\left(\chi,\Upsilon\right)\right]}{v^{n\varpi}} - \sum_{j=0}^{n-1} v^{(j-n)\varpi+2} \left(D_{\Upsilon}^{j\varpi}\Phi\right)$$

and $D_{\Upsilon}^{n\emptyset}\Phi = D_{\Upsilon}^{\emptyset}. D_{\Upsilon}^{\emptyset}. D_{\Upsilon}^{\emptyset}...D_{\Upsilon}^{\emptyset}(n-\text{times}).$

Theorem 1. Suppose that the function $E[\Phi(\Upsilon)] = \Psi(\upsilon)$ has *FPS representation as follows:*

$$\Psi(v) = \sum_{n=0}^{\infty} \aleph_n v^{n\overline{\omega}+2},$$
(32)

then we have $\aleph_n = (D_Y^{n\otimes}\Phi)(0)$, where $D_Y^{n\otimes}\Phi = D_Y^{\otimes}$, D_Y^{\otimes} , D_Y^{\otimes} , \dots , $D_Y^{\otimes}(n-times)$.

Proof: Consider new form of Taylor's series.

$$\Psi(v) = \aleph_0 v^2 + \aleph_1 v^{0+2} + \aleph_2 v^{20+2} + \aleph_3 v^{30+2} + \aleph_4 v^{40+2} + \cdots.$$
(33)

From the above equation, we have

$$\aleph_{1} = \frac{1}{v^{\omega+2}} \Psi(v) - \frac{1}{v^{\omega+2}} v^{2} \Psi(0) - \frac{1}{v^{\omega+2}} \aleph_{2} v^{2\omega+2} - \frac{1}{v^{\omega+2}} \aleph_{3} v^{3\omega+2} - \frac{1}{v^{\omega+2}} \aleph_{4} v^{4\omega+2} + \cdots.$$
(34)

Taking $v \longrightarrow 0$ on the above equation,

$$\aleph_{1} = \lim_{v \to 0} \frac{1}{v^{2}} \left(\frac{1}{v^{\varpi}} \Psi(v) - \frac{1}{v^{\varpi-2}} \Psi(0) \right).$$
(35)

By using Lemma 3,

where
$$\chi = (\chi_1, \chi_2, \chi_3, \dots, \chi_d) \in \mathfrak{R}^d, d \in N,$$
 (31)

$$\aleph_1 = \lim_{v \to 0} \frac{1}{v^2} \Big(E \Big[D_{\Upsilon}^{\omega} \Phi(\Upsilon) \Big](v) \Big).$$
(36)

By Lemma 2, the above equation becomes as

$$\aleph_1 = \left(D_{\Upsilon}^{\hat{\omega}} \Phi \right) (0). \tag{37}$$

Again from equation (32),

(0),

$$\aleph_2 = \frac{1}{v^2} \left(\frac{1}{v^{2\varpi}} \Psi(v) - \frac{1}{v^{2\varpi-2}} \Psi(0) - \frac{1}{v^{\varpi-2}} \aleph_1 \right) - \aleph_3 v^{\varpi}$$
$$- \aleph_4 v^{2\varpi} + \cdots.$$

(38)

Taking $v \longrightarrow 0$ on the last equation, utilizing Lemma 3, the above equation becomes as

$$\aleph_2 = \lim_{v \to 0} \frac{1}{v^2} \Big(E \Big[D_{\Upsilon}^{2\varpi} \Phi(\Upsilon) \Big](v) \Big).$$
(39)

By Lemma 2,

$$\aleph_2 = \left(D_{\Upsilon}^{2\omega} \Phi \right)(0). \tag{40}$$

Again from equation (32), we have

$$\aleph_{3} = \lim_{v \to 0} \frac{1}{v^{2}} \left(\frac{1}{v^{3\bar{\omega}}} \Psi(v) - \Psi(0) \frac{1}{v^{3\bar{\omega}-2}} - \aleph_{1} \frac{1}{v^{2\bar{\omega}}} - \aleph_{2} \frac{1}{v^{\bar{\omega}-2}} \right).$$
(41)

By Lemma 3,

$$\aleph_3 = \lim_{v \to 0} \left(\frac{1}{v^2} E \left[D_{\Upsilon}^{3\omega} \Phi(\Upsilon) \right](v) \right).$$
(42)

By Lemma 2, the last equation becomes as

$$\aleph_3 = \left(D_{\Upsilon}^{30} \Phi \right)(0). \tag{43}$$

In the same manner, we can obtain the following form by making generalization:

$$\aleph_n = \left(D_{\Upsilon}^{n\omega} \Phi \right)(0). \tag{44}$$

This completes the proof of the theorem. \Box

Remark 4. For multiple Taylor's series, the proved result becomes as follows:

$$\aleph_n(\chi) = \left(D_{\Upsilon}^{n\omega}\Phi\right)(0), \quad where \ \chi = \left(\chi_1, \chi_2, \chi_3, \dots, \chi_d\right) \in \Re^d, d \in N.$$
(45)

The following theorem describes and determines the conditions for convergence of the new form of Taylor's formula that are introduced in Lemma 1.

Theorem 2. Let $\Psi(v) = E[\Phi(\Upsilon)]$ be represented as the new form of fractional Taylor's formula as in Elzaki transform:

$$\Psi(v) = \sum_{n=0}^{\infty} \aleph_n v^{n\omega+2}, \qquad (46)$$

$$\left|\frac{1}{v^2}E\left[D_{\Upsilon}^{(n+1)\mathfrak{d}}\Phi(\Upsilon)\right]\right| \le T.$$
(47)

Then, the remainder $R_n(v)$ of the new form of fractional Taylor's formula satisfies the following inequality:

$$\left|R_{n}\left(v\right)\right| \le v^{(n+1)\varpi+2}T.$$
(48)

Proof: Consider the following:

$$\Psi_{n}(v) = \aleph_{0}v^{2} + \aleph_{1}v^{\omega+2} + \aleph_{2}v^{2\omega+2} + \aleph_{3}v^{3\omega+2} + \cdots + \aleph_{n}v^{n\omega+2}.$$
(49)

From equations (46) and (49), we get

$$R_n(v) = \Psi(v) - \sum_{k=0}^n \aleph_k v^{k\bar{\omega}+2}.$$
 (50)

By Theorem 1,

$$R_{n}(v) = \Psi(v) - \sum_{k=0}^{n} v^{k\bar{\omega}+2} \left(D_{Y}^{k\bar{\omega}} \Phi \right)(0),$$

$$\frac{1}{v^{(n+1)\bar{\omega}+2}} R_{n}(v) = \frac{1}{v^{2}} \left(\frac{1}{v^{(n+1)\bar{\omega}}} \Psi(v) - \sum_{k=0}^{n} \frac{1}{v^{(n+1-k)\bar{\omega}-2}} \left(D_{Y}^{k\bar{\omega}} \Phi \right)(0) \right).$$
(51)

By Lemma 3,

$$\frac{1}{v^{(n+1)\bar{\omega}+2}}R_n(v) = \frac{1}{v^2}E\left[D_{\Upsilon}^{(n+1)\bar{\omega}}\Phi(\Upsilon)\right],$$

$$\left|\frac{1}{v^{(n+1)\bar{\omega}+2}}R_n(v)\right| = \left|\frac{1}{v^2}E\left[D_{\Upsilon}^{(n+1)\bar{\omega}}\Phi(\Upsilon)\right]\right|.$$
(52)

By the given assumption, the above equation becomes as

$$-v^{(n+1)\bar{\omega}+2}T \le R_n(v) \le v^{(n+1)\bar{\omega}+2}T,$$
(53)

 $|R_n(v)| \le v^{(n+1)\overline{\omega}+2}T$. Hence, the required result is proved.

3. Demonstrating the ERPSM for the CRDEs

We exploit our novel ERPSM to originate the results of the linear and nonlinear CRDEs. The foremost set of rules of this method for resolving the CRDEs can be accumulated by the following steps: employing the Elzaki transform to CRDE and then deploying the novel form of Taylor's series to introduce the solution of CRDE in the novel space. The coefficients of this series are established with a new idea. At the end, employing the inverse Elzaki transform to achieve the solution of the problem in the actual space.

3.1. Elzaki Residual Power Series Solutions for the CRDEs. In this subsection, we systematized the stages for conquering the Elzaki residual power series solution for the linear and nonlinear CRDE by the following procedure.

Step 1. Rewriting equation (2) as demonstrated:

$$\frac{\partial^{\omega}\Phi(\chi,\Upsilon)}{\partial\Upsilon^{\omega}} - \lambda \frac{\partial^{2}\Phi(\chi,\Upsilon)}{\partial\chi^{2}} - z(\chi,\Upsilon)\Phi(\chi,\Upsilon) = 0.$$
(54)

Step 2. Manipulating Elzaki transform at both sides of equation (54), we get in this way

$$E\left[\frac{\partial^{\omega}\Phi(\chi,\Upsilon)}{\partial\Upsilon^{\omega}}\right] - \lambda E\left[\frac{\partial^{2}\Phi(\chi,\Upsilon)}{\partial\chi^{2}}\right] - E\left[E^{-1}\left[Z(\chi,v)\right]E^{-1}\left[\Psi(\chi,v)\right]\right] = 0,$$
(55)

if

where

$$E^{-1}[Z(\chi, v)] = z(\chi, \Upsilon),$$

$$E^{-1}[\Psi(\chi, v)] = \Phi(\chi, \Upsilon),$$

$$E\left[\frac{\partial^{\varpi}\Phi(\chi, \Upsilon)}{\partial \Upsilon^{\varpi}}\right] = \frac{\Psi(\chi, v)}{v^{\varpi}} - v^{-\varpi+2}g(\chi).$$
(56)

So, we get the following form:

$$\Psi(\chi, v) = v^2 g(\chi) + \lambda v^{\hat{\omega}} D_{\chi\chi} \Psi(\chi, v) + v^{\hat{\omega}} E \Big[E^{-1} [Z(\chi, v)] E^{-1} [\Psi(\chi, v)] \Big].$$
(57)

Step 3. Considering the solution of equation (57) as the following:

$$\Psi(\chi, v) = \sum_{n=0}^{\infty} \aleph_n(\chi) v^{2+n\mathfrak{d}}.$$
 (58)

Step 4. Setting $\aleph_0(\chi) = \lim_{v \to \infty} (1/v^2) \Psi(\chi, v) = \Phi(\chi, 0)$. Step 5. Establishing the *k*th-truncated series of $\Psi(\chi, v)$ as

$$\Psi_{k}(\chi, v) = \sum_{n=0}^{k} \aleph_{n}(\chi) v^{2+n\omega},$$

$$\aleph_{0} = \lim_{v \longrightarrow 0} \frac{1}{v^{2}} \Psi(\chi, v),$$
(59)

$$\Psi_{k}(\chi, v) = \aleph_{0} v^{2} + \sum_{n=1}^{k} \aleph_{n}(\chi) v^{2+n\omega}.$$

Step 6. Considering the Elzaki residual function (ERF) of equation (57) and the *k*th-truncated ERF separately such that

$$E\operatorname{Res}(\chi, v) = \Psi(\chi, v) - v^{2}g(\chi) - \lambda v^{\omega}D_{\chi\chi}\Psi(\chi, v)$$
$$- v^{\omega}E\left[E^{-1}[Z(\chi, v)]E^{-1}[\Psi(\chi, v)]\right],$$
$$E\operatorname{Res}_{k}(\chi, v) = \Psi_{k}(\chi, v) - v^{2}g(\chi) - \lambda v^{\omega}D_{\chi\chi}\Psi_{k}(\chi, v)$$
$$- v^{\omega}E\left[E^{-1}[Z(\chi, v)]E^{-1}[\Psi_{k}(\chi, v)]\right].$$
(60)

Step 7. Replacing the series form of $\Psi_k(\chi, v)$ into equation (60).

Step 8. Dividing at both sides of equation (60) with v^{kv+2} as follows:

$$\frac{1}{v^{2+k\overline{\omega}}} E \operatorname{Res}_{k}(\chi, v) = \frac{1}{v^{2+k\overline{\omega}}} \Psi_{k}(\chi, v) - \frac{1}{v^{2+k\overline{\omega}}} v^{2} g(\chi)$$
$$- \lambda \frac{1}{v^{2+k\overline{\omega}}} v^{\overline{\omega}} D_{\chi\chi} \Psi_{k}(\chi, v)$$
$$- \frac{1}{v^{2+k\overline{\omega}}} v^{\overline{\omega}} E \left[E^{-1} [Z(\chi, v)] E^{-1} [\Psi_{k}(\chi, v)] \right].$$
(61)

Step 9. Taking limit at both sides of equation (61).

$$\lim_{v \to 0} \frac{1}{v^{2+k\bar{\omega}}} E \operatorname{Res}_{k}(\chi, v) = \lim_{v \to 0} \frac{1}{v^{2+k\bar{\omega}}} \Psi_{k}(\chi, v) - \lim_{v \to 0} \frac{1}{v^{2+k\bar{\omega}}} v^{2} g(\chi) - \lambda \lim_{v \to 0} \frac{1}{v^{2+k\bar{\omega}}} v^{\bar{\omega}} D_{\chi\chi} \Psi_{k}(\chi, v) - \lim_{v \to 0} \frac{1}{v^{2+k\bar{\omega}}} v^{\bar{\omega}} E \Big[E^{-1} [Z(\chi, v)] E^{-1} [\Psi_{k}(\chi, v)] \Big].$$
(62)

Step 10. Solving the following equation for $\aleph_n(\chi)$:

$$\lim_{v \to 0} \left(\frac{1}{v^{k\bar{\omega}+2}} E \operatorname{Res}_{k}(\chi, v) \right) = 0, \quad k = 1, 2, 3, \dots$$
(63)

Step 11. Replacing the attained values of $\aleph_n(\chi)$ into *k*th-truncated series of $\Psi(\chi, v)$ to get the *k*th-approximate solution of equation (57).

Step 12. Manipulating the inverse Elzaki transform on $\Psi_k(\chi, v)$ to attain the *k*th-approximate solution of $\Phi_k(\chi, \Upsilon)$ in the real space.

3.2. Applications to Linear and Nonlinear CRDEs. In this subsection, we consider three main problems of CRDEs to illustrate the execution and capability of ERPSM.

3.2.1. Approximate and Closed Form Solutions of Linear CRDEs. Two applications are considered for linear CRDEs.

Problem 1. Consider the time-fractional linear CRDE [19].

$$D_{\Upsilon}^{\varpi}\Phi(\chi,\Upsilon) = \Phi_{\chi\chi}(\chi,\Upsilon) - \Phi(\chi,\Upsilon), \quad \chi, \Upsilon \ge 0, \ 0 < \varpi \le 1.$$
(64)

Subject to initial condition,

$$\Phi(\chi,0) = e^{-\chi} + \chi. \tag{65}$$

Solution. Utilizing Elzaki transform on equation (64),

$$E\left[D_{\Upsilon}^{0}\Phi\left(\chi,\Upsilon\right)\right] = E\left[\Phi_{\chi\chi}\left(\chi,\Upsilon\right)\right] - E\left[\Phi\left(\chi,\Upsilon\right)\right],\tag{66}$$

where $E[\Phi(\chi, \Upsilon)] = \Psi(\chi, v)$,

$$E\left[D_{\Upsilon}^{0}\Phi\left(\chi,\Upsilon\right)\right] = \frac{\Psi(\chi,\upsilon)}{\upsilon^{0}} - \upsilon^{-\omega+2}\Phi\left(\chi,0\right),\tag{67}$$

so equation (66) becomes as

$$\Psi(\chi, v) = v^2 \Phi(\chi, 0) + v^{\omega} D_{\chi\chi} \Psi(\chi, v) - v^{\omega} \Psi(\chi, v).$$
(68)

Initiate a series solution to the algebraic equation (68). Hence, presume that the expansion of $\Psi(\chi, v)$ is the following:

$$\Psi(\chi, v) = \sum_{n=0}^{\infty} \aleph_n(\chi) v^{2+n\emptyset}.$$
(69)

Assume that $\Psi(\chi, v)$ has the *k*th-truncated series as

$$\Psi_k(\chi, v) = \sum_{n=0}^k \aleph_n(\chi) v^{2+n\omega}.$$
(70)

By Lemma 2, we have

$$\lim_{v \to 0} \frac{1}{v^2} \Psi(\chi, v) = \Phi(\chi, 0) = e^{-\chi} + \chi.$$
(71)

The *k*th-truncated series becomes as follows:

$$\Psi_k(\chi, v) = v^2 \left(e^{-\chi} + \chi \right) + \sum_{n=1}^k \aleph_n(\chi) v^{2+n\varpi}.$$
 (72)

The ERF of the algebraic equation (68) is described as

$$E\operatorname{Res}\left(\chi,v\right) = \Psi\left(\chi,v\right) - v^{2}\left(e^{-\chi} + \chi\right) - v^{\tilde{\omega}}D_{\chi\chi}\Psi\left(\chi,v\right) + v^{\tilde{\omega}}\Psi\left(\chi,v\right).$$
(73)

Furthermore, *k*th-truncated ERF of the algebraic equation (69) is explained as follows:

$$E\operatorname{Res}_{k}(\chi, v) = \Psi_{k}(\chi, v) - v^{2} \left(e^{-\chi} + \chi\right) - v^{\varpi} D_{\chi\chi} \Psi_{k}(\chi, v) + v^{\varpi} \Psi_{k}(\chi, v).$$
(74)

By utilizing equations (72) and (74), we get undefined coefficients in the following form:

$$\begin{split} \aleph_{1}(\chi) &= -\chi, \\ \aleph_{2}(\chi) &= \chi, \\ \aleph_{3}(\chi) &= -\chi, \\ \aleph_{4}(\chi) &= \chi, \\ \aleph_{5}(\chi) &= -\chi. \end{split}$$
(75)

So, we get the 5th approximate solution of Elzaki transform of equation (68).

$$\Psi_{5}(\chi, v) = \frac{e^{-\chi} + \chi}{v^{2}} - \frac{\chi}{v^{2+\omega}} + \frac{\chi}{v^{2+2\omega}} - \frac{\chi}{v^{2+3\omega}} + \frac{\chi}{v^{2+4\omega}} - \frac{\chi}{v^{2+5\omega}}.$$
(76)

Operating inverse Elzaki transform on both sides of equation (76), we get the 5^{th} approximate solution of equation (64).

$$\Phi_{5}(\chi,\Upsilon) = e^{-\chi} + \chi \left(1 - \frac{\Upsilon^{0}}{\Gamma(0+1)} + \frac{\Upsilon^{20}}{\Gamma(20+1)} - \frac{\Upsilon^{30}}{\Gamma(30+1)} + \frac{\Upsilon^{40}}{\Gamma(40+1)} - \frac{\Upsilon^{50}}{\Gamma(50+1)}\right).$$
(77)

When $\omega = 1$, equation (77) becomes as

$$\Phi_5(\chi,\Upsilon) = e^{-\chi} + \chi \left(1 - \frac{\Upsilon}{1!} + \frac{\Upsilon^2}{2!} - \frac{\Upsilon^3}{3!} + \frac{\Upsilon^4}{4!} - \frac{\Upsilon^5}{5!} \right).$$
(78)

Equation (78) is coinciding with the six terms of the expansion of the exact solution $\Phi(\chi, \Upsilon) = e^{-\chi} + \chi e^{-\Upsilon}$.

Table 1 demonstrates the values of absolute error of the 5th order approximate and exact solutions at $\bar{\omega} = 1$ when $\chi = 1$ which support the capability and exactness of the novel technique.

Figure 1 displays the evaluations of exact solution at $\varpi = 1$ and the 5th approximate solution of Problem 1, at $\chi = 1$, for several values of Υ and ϖ . Figure 1 confirms that when values of ϖ approach to "1," the approximate solution approaches to the exact solution, which approves the efficacy and correctness of the new method. Moreover, the approximate

solution overlaps with the exact solution at $\omega = 1$ and this once more ratifies the usefulness and correctness of the ERPSM.

Problem 2. Consider the time-fractional linear CRDE [20],

$$D_{\Upsilon}^{a}\Phi\left(\chi,\Upsilon\right) = \Phi_{\chi\chi}\left(\chi,\Upsilon\right) - \left(1 + 4\chi^{2}\right)\Phi\left(\chi,\Upsilon\right),$$

$$\chi,\Upsilon \ge 0, \ 0 < a \le 1.$$
(79)

With the initial condition,

$$\Phi(\chi,0) = e^{\chi^2}.$$
 (80)

Solution. Manipulating Elzaki transform on equation (79),

TABLE 1: Absolute error of ERPS results.

Υ	Exact solution	Approximate solution	Absolute error
0	2.71828182846	2.71828182846	0
0.06	2.88637098927	2.88637098909	$1.7766543792 \times 10 - 10$
0.12	3.06485420329	3.06485419182	$1.1469452055 \times 10 - 8$
0.18	3.25437420289	3.2543740711	$1.3178722424 \times 10-7$
0.24	3.45561346476	3.45561271778	$7.4698766506 \times 10-7$
0.30	3.66929666762	3.66929379283	0.00000287478963168
0.36	3.8961933018	3.89618464113	0.00000866066429372
0.42	4.13712044025	4.13709840516	0.0000220350955411
0.48	4.39294568092	4.39289613873	0.044093686996



FIGURE 1: Evaluation of closed form and approximate consequences of Problem 1.

$$\begin{split} E\left[D_{Y}^{\emptyset}\Phi\left(\chi,\Upsilon\right)\right] &= E\left[\Phi_{\chi\chi}\left(\chi,\Upsilon\right)\right] - \left(1 + 4\chi^{2}\right)E\left[\Phi\left(\chi,\Upsilon\right)\right],\\ \Psi\left(\chi,\upsilon\right) &= \upsilon^{2}e^{\chi^{2}} + \upsilon^{\emptyset}D_{\chi\chi}\Psi\left(\chi,\upsilon\right) - \upsilon^{\emptyset}\left(1 + 4\chi^{2}\right)\Psi\left(\chi,\upsilon\right), \end{split}$$

$$\end{split}$$

$$\tag{81}$$

where

$$E\left[D_{\Upsilon}^{\mathbb{Q}}\Phi\left(\chi,\Upsilon\right)\right] = \frac{\Psi\left(\chi,\upsilon\right)}{\upsilon^{\mathbb{Q}}} - \upsilon^{-\overline{\omega}+2}e^{\chi^{2}},$$

$$E\left[\Phi\left(\chi,\Upsilon\right)\right] = \Psi\left(\chi,\upsilon\right).$$
(82)

Now, establishing a series solution of equation (81), consequently assume that $\Psi(\chi, v)$ has the expansion as follows:

$$\Psi(\chi, v) = \sum_{n=0}^{\infty} \aleph_n(\chi) v^{2+n\emptyset}.$$
(83)

The *k*th-truncated series $\Psi(\chi, v)$ is as follows:

$$\Psi_k(\chi, v) = \sum_{n=0}^k \aleph_n(\chi) v^{2+n\overline{\omega}}.$$
(84)

By Lemma 2, we have

$$\lim_{v \to 0} \frac{1}{v^2} \Psi(\chi, v) = \Phi(\chi, 0) = e^{\chi^2}.$$
 (85)

So, equation (84) becomes as

$$\Psi_k(\chi, v) = e^{\chi^2} v^2 + \sum_{n=1}^k \aleph_n(\chi) v^{2+n\bar{\omega}}.$$
 (86)

The ERF of equation (81) is defined as

$$E\operatorname{Res}(\chi, v) = \Psi(\chi, v) - v^{2} e^{\chi^{2}} - v^{\varpi} D_{\chi\chi} \Psi(\chi, v) + v^{\varpi} (1 + 4\chi^{2}) \Psi(\chi, v).$$
(87)

The *k*th-ERF is as follows:

$$ERes_{k}(\chi, v) = \Psi_{k}(\chi, v) - v^{2}e^{\chi^{2}} - v^{\varpi}D_{\chi\chi}\Psi_{k}(\chi, v) + v^{\varpi}(1 + 4\chi^{2})\Psi_{k}(\chi, v).$$
(88)

To find unspecified coefficients using equations (86) and (88), so we have

$$\begin{split} \aleph_1(\chi) &= e^{\chi^2}, \\ \aleph_2(\chi) &= e^{\chi^2}, \\ \aleph_3(\chi) &= e^{\chi^2}, \\ \aleph_4(\chi) &= e^{\chi^2}, \\ \aleph_5(\chi) &= e^{\chi^2}. \end{split}$$
(89)

The 5th approximate solution of equation (81) in Elzaki transform form is

$$\Psi_{5}(\chi, v) = e^{\chi^{2}} \left(\frac{1}{v^{2}} + \frac{1}{v^{2+\bar{\omega}}} + \frac{1}{v^{2+2\bar{\omega}}} + \frac{1}{v^{2+3\bar{\omega}}} + \frac{1}{v^{2+4\bar{\omega}}} + \frac{1}{v^{2+5\bar{\omega}}} \right).$$
(90)

By applying inverse Elzaki transform on equation (90), we get the 5^{th} approximate solution of equation (79) as follows:

$$\Phi_{5}(\chi,\Upsilon) = e^{\chi^{2}} \left(1 + \frac{\Upsilon^{\varpi}}{\Gamma(\varpi+1)} + \frac{\Upsilon^{2\varpi}}{\Gamma(2\varpi+1)} + \frac{\Upsilon^{3\varpi}}{\Gamma(3\varpi+1)} + \frac{\Upsilon^{4\varpi}}{\Gamma(4\varpi+1)} + \frac{\Upsilon^{5\varpi}}{\Gamma(5\varpi+1)}\right).$$
(91)

When $\omega = 1$, equation (91) becomes as

$$\Phi_5(\chi,\Upsilon) = e^{\chi^2} \left(1 + \frac{\Upsilon}{1!} + \frac{\Upsilon^2}{2!} + \frac{\Upsilon^3}{3!} + \frac{\Upsilon^4}{4!} + \frac{\Upsilon^5}{5!} \right).$$
(92)

Equation (92) represents the first six terms of the expansion of $e^{\chi^2+\Upsilon}$, so closed form solution of equation (79) is $e^{\chi^2+\Upsilon}$.

Table 2 demonstrates the values of absolute error of the 5th order approximate and exact solutions at $\omega = 1$ when $\chi = 1$, which support the capability and accuracy of the new technique.

Figure 2 demonstrates the exploits of exact solution at $\varpi = 1$ and the 5th approximate solution of Problem 2, when $\chi = 1$ for numerous values of Y and ϖ . The figure endorses that when values of ϖ approach to "1," the approximate solution approaches to exact solution, which supports the ability and precision of the new method. Moreover, the approximate solution overlaps with the exact solution at $\varpi = 1$ and this once more ratifies the usefulness and correctness of the ERPSM.

3.2.2. Approximate and Closed Form Solutions of Nonlinear CRDEs

Problem 3. Consider the nonlinear time-fractional CRDE [21],

$$D_{Y}^{\omega}\Phi(\chi,\Upsilon) = \Phi_{\chi\chi}(\chi,\Upsilon) - \Phi_{\chi}(\chi,\Upsilon) + \Phi(\chi,\Upsilon)\Phi_{\chi\chi}(\chi,\Upsilon)$$
$$-\Phi^{2}(\chi,\Upsilon) + \Phi(\chi,\Upsilon), \quad \chi,\Upsilon \ge 0, \ 0 < \omega \le 1,$$
(93)

With the initial condition,

$$\Phi\left(\chi,0\right) = e^{\chi}.\tag{94}$$

Solution. By applying Elzaki transform on equation (93), we get

$$\Psi(\chi, v) = v^{2} e^{\chi} + v^{\tilde{\omega}} D_{\chi\chi} \Psi(\chi, v) - v^{\tilde{\omega}} D_{\chi} \Psi(\chi, v) + v^{\tilde{\omega}} E \left[E^{-1} \left[\Psi(\chi, v) \right] D_{\chi\chi} E^{-1} \left[\Psi(\chi, v) \right] \right] - v^{\tilde{\omega}} E \left[\left[E^{-1} \Psi(\chi, v) \right]^{2} \right] + v^{\tilde{\omega}} \Psi(\chi, v).$$
(95)

Here,

$$E\left[\Phi_{\chi\chi}(\chi,\Upsilon)\right] = D_{\chi\chi}\Psi(\chi,v),$$

$$E\left[\Phi_{\chi}(\chi,\Upsilon)\right] = D_{\chi}\Psi(\chi,v),$$

$$\Phi(\chi,\Upsilon) = E^{-1}[\Psi(\chi,v)],$$

$$\Phi^{2}(\chi,\Upsilon) = \left[E^{-1}[\Psi(\chi,v)]\right]^{2},$$

$$\Phi_{\chi\chi}(\chi,\Upsilon) = D_{\chi\chi}E^{-1}[\Psi(\chi,v)],$$

$$E\left[D_{\Upsilon}^{\omega}\Phi(\chi,\Upsilon)\right] = \frac{\Psi(\chi,v)}{v^{\omega}} - v^{-\omega+2}e^{\chi}.$$
(96)

Define a series solution of equation (95) as follows:

$$\Psi(\chi, \upsilon) = \sum_{n=0}^{\infty} \aleph_n(\chi) \upsilon^{2+n\varpi}.$$
(97)

The kth-truncated series is

$$\Psi_k(\chi, v) = \sum_{n=0}^k \aleph_n(\chi) v^{2+n\omega}.$$
(98)

By Lemma 2,

$$\lim_{v \to 0} \frac{1}{v^2} \Psi(\chi, v) = \Phi(\chi, 0) = e^{\chi}.$$
 (99)

Therefore, the kth-truncated series becomes as

$$\Psi_k(\chi, v) = e^{\chi} v^2 + \sum_{n=1}^k \aleph_n(\chi) v^{2+n\omega}.$$
 (100)

Now, define ERF in the following form:

$$E\operatorname{Res}\left(\chi,v\right) = \Psi\left(\chi,v\right) - v^{2}e^{\chi} - v^{\varpi}D_{\chi\chi}\Psi\left(\chi,v\right) + v^{\varpi}D_{\chi}\Psi\left(\chi,v\right) - v^{\varpi}E\left[E^{-1}\left[\Psi\left(\chi,v\right)\right]D_{\chi\chi}E^{-1}\left[\Psi\left(\chi,v\right)\right]\right] + v^{\varpi}E\left[\left[E^{-1}\Psi\left(\chi,v\right)\right]^{2}\right] - v^{\varpi}\Psi\left(\chi,v\right).$$
(101)

The *k*th-truncated ERF is

Ŷ	Exact solution	Approximate solution	Absolute error
0.00	1.36787944117	1.36787944117	0.0
0.06	1.30964397476	1.30964397469	$7 \times 10 - 11$
0.12	1.25479987789	1.25479987381	$4.08 \times 10 - 9$
0.18	1.20314965258	1.20314960653	$4.605 \times 10 - 8$
0.24	1.15450730224	1.15450704565	$2.5659 \times 10 - 7$
0.30	1.10869766185	1.10869669117	$9.7068 \times 10-7$
0.36	1.06555576724	1.06555289269	0.00000287455
0.42	1.02492626099	1.02491907181	0.00000718918
0.48	0.98666283297	0.98664694453	0.000015888447

TABLE 2: Absolute error of ERPS results.



FIGURE 2: The behavior of exact and approximate outcomes of Problem 2.

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$$ERes_{k}(\chi, v) = \Psi_{k}(\chi, v) - v^{2}e^{\chi} - v^{\tilde{\omega}}D_{\chi\chi}\Psi_{k}(\chi, v)$$
$$+ v^{\tilde{\omega}}D_{\chi}\Psi_{k}(\chi, v)$$
$$- v^{\tilde{\omega}}E\left[E^{-1}[\Psi_{k}(\chi, v)]D_{\chi\chi}E^{-1}[\Psi_{k}(\chi, v)]\right]$$
$$+ v^{\tilde{\omega}}E\left[\left[E^{-1}\Psi_{k}(\chi, v)\right]^{2}\right] - v^{\tilde{\omega}}\Psi_{k}(\chi, v).$$
(102)

~

The undefined coefficients are determined in the following form by utilizing equations (100) and (102).

$$\begin{split} \aleph_1(\chi) &= e^{\chi}, \\ \aleph_2(\chi) &= e^{\chi}, \\ \aleph_3(\chi) &= e^{\chi}, \\ \aleph_4(\chi) &= e^{\chi}, \\ \aleph_5(\chi) &= e^{\chi}. \end{split}$$
(103)

The 5th approximate solution of equation (95) is given as

`

$$\Psi_{5}(\chi, v) = e^{\chi} \left(\frac{1}{v^{2}} + \frac{1}{v^{2+\bar{\omega}}} + \frac{1}{v^{2+2\bar{\omega}}} + \frac{1}{v^{2+3\bar{\omega}}} + \frac{1}{v^{2+4\bar{\omega}}} + \frac{1}{v^{2+5\bar{\omega}}} \right).$$
(104)

By applying inverse Elzaki transform on the above equation, we get the 5^{th} approximate solution of equation (93).

$$\Phi_{5}(\chi,\Upsilon) = e^{\chi} \left(1 + \frac{\Upsilon^{\varpi}}{\Gamma(\varpi+1)} + \frac{\Upsilon^{2\varpi}}{\Gamma(2\varpi+1)} + \frac{\Upsilon^{3\varpi}}{\Gamma(3\varpi+1)} + \frac{\Upsilon^{4\varpi}}{\Gamma(4\varpi+1)} + \frac{\Upsilon^{5\varpi}}{\Gamma(4\varpi+1)}\right).$$
(105)

For $\omega = 1$, the last equation becomes as

Υ	Exact solution	Approximate solution	Absolute error
0.00	1.36787944117	1.36787944117	0.0
0.06	1.30964397476	1.30964397469	$7 \times 10 - 11$
0.12	1.25479987789	1.25479987381	$4.08 \times 10 - 9$
0.18	1.20314965258	1.20314960653	$4.605 \times 10 - 8$
0.24	1.15450730224	1.15450704565	$2.5659 \times 10 - 7$
0.30	1.10869766185	1.10869669117	$9.7068 \times 10 - 7$
0.36	1.06555576724	1.06555289269	0.00000287455
0.42	1.02492626099	1.02491907181	0.00000718918
0.48	0.98666283297	0.98664694453	0.000015888447

TABLE 3: Absolute error of ERPS results.



FIGURE 3: 2D plot of exact and approximate solution of Problem 3.

$$\Phi_5(\chi,\Upsilon) = e^{\chi} \left(1 + \frac{\Upsilon}{1!} + \frac{\Upsilon^2}{2!} + \frac{\Upsilon^3}{3!} + \frac{\Upsilon^4}{4!} + \frac{\Upsilon^5}{5!} \right).$$
(106)

Equation (106) coincides with the 1st six terms of the expansion of $e^{\chi+\Upsilon}$, therefore exact solution of equation (93) is $e^{\chi+\Upsilon}$.

Table 3 demonstrates the values of absolute error of the 5th order approximate and exact results at $\omega = 1$, and $\chi = 1$, which support the ability and accuracy of the novel technique.

Figure 3 establishes the actions of exact solution at $\omega = 1$ and the 5th approximate solution of Problem 3, when $\chi = 1$ for certain values of Y and ω . The figure recommends that when values of ω approach to "1," the approximate solution approaches to exact solution, which supports the capability and exactness of the new method. Moreover, the approximate solution overlaps with the exact solution at $\omega = 1$ and this once more ratifies the usefulness and correctness of the ERPSM.

4. Conclusions

There are enormous number of numerical and analytical methods for resolving the DEs; there are numerous methods that have superiority over the others. Few of them are precise and operative, but they necessitate mathematical operations that can be problematic and elongated. Our novel method, ERPSM, is considered by accurateness, rapidity, and effortlessness in finding exact and approximate solutions to DEs.

To study the efficiency and reliability of ERPSM for PDEs, absolute errors of three applications are scrutinized. Consequences verify that our novel technique is simple, accurate, applicable, and efficient. The recommended techniques offered us an effortless and quick technique to perceive the coefficients of the suggested series to be a solution to the equation. Dissimilar to the traditional RPS method, while establishing the coefficients for a series, it is required to compute the fractional derivative every time, while ERPSM only requires the concept of the limit at zero in establishing the coefficients for the series.

The gain of the ERPSM is that it decreases considerably the numerical calculations to construct the consequences for this category of equations related to existing methods, for instance, the differential transform method (DTM), perturbation method, and Adomian decomposition method (ADM). Consequently, we can accomplish that the ERPSM is effortless, effective, and practical for solving numerous further fractional order PDEs.

Data Availability

No data were used to support this study.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

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Research Article

Dynamic Response Analysis of a Forced Fractional Viscoelastic Beam*

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In this paper, dynamic response analysis of a forced fractional viscoelastic beam under moving external load is studied. The beauty of this study is that the effect of values of fractional order, the effect of internal damping, and the effect of intensity value of the moving force load on the dynamic response of the beam are analyzed. Constitutive equations for fractional order viscoelastic beam are constructed in the manner of Euler–Bernoulli beam theory. Solution of the fractional beam system is obtained by using Bernoulli collocation method. Obtained results are presented in the tables and graphical forms for two different beam systems, which are polybutadiene beam and butyl B252 beam.

1. Introduction

Theory and applications of beams are very important research area due to its wide usage areas in applied sciences. Especially after starting the space adventure of the mankind, the demand to more resistant structures has great imporgenerally modeled based tance. Beams are on Euler-Bernoulli beam theory, which is called classical beam theory. The background of beam theory goes on Newton's second law and some different aspects of beams, such as modeling, analysis of bending-buckling, and reinforcement and control, are hot topics of research papers since the beginning of the nineteenth century. The books can provide a general overview about the Euler-Bernoulli beam theory, please see [1-3]. Some important studies related to beams modeled in the sense of classical beam theory are also summarized as follows, but not limited to [4-15]. The beam systems in [1–15] have the integer order derivatives of the state function. In the beginning of 1930s, fractional derivative was introduced for describing the constitutive relation of some beam materials [16], and after 1980s, since fractional order equations have good memory and can be used to describe material properties more accurately with fewer

parameters, they are considered to be good mathematical models for describing the dynamic mechanical behavior of materials [17]. In [18], the dynamic behavior of the thin plates resting on a fractionally damped viscoelastic foundation subjected to a moving point load is investigated and results show that the damping of the foundation system increases with increasing the order of the fractional derivative, which leads to a decrease in the dynamic response. In [19], the dynamic response spectra of fractionally damped viscoelastic beams subjected to concentrated moving load are presented and results reveal that with an increase in the order of the fractional derivative, the system damping of the system increases and the dynamic amplification factor (DAF) decreases, especially in the dynamic zone of the sweep parameter. In [20], the precise integration method (PIM) is extended to numerically integrate the equation of motion with fractional terms, which offers high accuracy and obtained numerical results indicate the viscoelastic dampers can enhance the seismic performance of structures significantly. In [21], the nonstationary free vibration and nonlinear dynamic behavior of the viscoelastic nanoplates are analyzed. Obtained results show that the viscoelastic modelbased vibration is nonstationary unlike the elastic model. Moreover, the damping mechanism of the viscoelasticity is amplitude dependent and the contribution of the viscoelastic damping terms at higher forcing conditions becomes noticeable. On the other hand, several numerical methods are developed and employed for better analyzing the fractional mechanical systems. Widely used methods for fractional systems are finite element method [22], Galerkin method [23], variational iteration method [24], and multiscale method [25, 26]. Especially, papers existing in the literature, which include a solution method for analyzing the dynamic response of a fractional order beam system, can be shortly listed as [19, 27–30]. In [19], the authors combined Galerkin method and Newton-Raphson method for analyzing the vibration of a fractional beam equation and they compared the results for only seeing the effects of fractional or integer derivatives. In [27], the author considered the dynamic response analyzing of a fractional order viscoelastic beam by means of green function method. In [27], the author only compared the results based on changes on the fractional derivative between (0, 1). In [28], the authors employed the Adomian decomposition method for solving a fractional beam equation and they only observed the effect of the order of fractional derivative. In [29], the authors used the dynamic green function method for analyzing the dynamic response in a fractional beam equation and the beam equation does not include the damping term. Results are simulated for only indicating the effects of order of fractional derivative. In [30], the author employed the green function method for a fractional viscoelastic beam system subjected to a base excitation. After obtaining the solution, the author compared the results corresponding to different fractional order derivative. By comparing the present study with the studies existing in the literature, objectives of the present study are expressed as follows:

- (i) In this paper, Bernoulli collocation method is firstly employed for analyzing the fractional viscoelastic beam equation. In the literature, especially for the fractional beam systems, green function method, Galerkin method, Newton-Raphson method, Adomian decomposition method, and Bernoulli collocation method in this paper were used, but by comparing these five methods, it is clear that Bernoulli collocation method is new and has less computational process and less work.
- (ii) In the literature, the authors only considered and discussed the effects of order of fractional derivative on the dynamic response. But, we discussed both the effects of the order of fractional order derivative and the effects of damping coefficient term and the effect of density of moving force load. So, it is said that the present study has wider perspective than other studies.
- (iii) Also, in the literature, results are obtained for one beam system. In this paper, effects of order of fractional derivative, effects of damping coefficient term, and the effect of density of moving force load are observed and compared for two different beam

systems which are polybutadiene beam and butyl B252 beam.

For theoretical and experimental review about the fractional Euler-Bernoulli beams, please see [31]. Specifically, in the present paper, displacement analysis of a forced fractional viscoelastic beam is studied. External moving force load perfectly moves on the beam with the velocity v(t)from the left edge to the right edge of the beam. The solution of the fractional beam system is obtained by means of Bernoulli collocation method. The main advantage of the Bernoulli collocation method is that employing the Bernoulli polynomials is easier than Chebyshev, Bessel polynomials, and Haar wavelets [32-34]. These advantages of Bernoulli polynomials provide us for obtaining the solution by making less computational process in shorter time. In the step of employing the Bernoulli collocation method, some external moving force loads having different load intensities are considered and also the effects of internal damping and fractional order of the derivative are searched for a fractional beam system. In the simulations, two different beam systems, which are polybutadiene beam and butyl B252 beam, are taken into account for being compared each other in the aspects of internal damping effects and resistance to effect of external moving force. Comparison results of the beam systems are presented in tables and graphics. The rest of the paper is organized as follows: in the next section, definition of the displacement analysis problem for a fractional viscoelastic beam is presented and scheme of the beam is overviewed. In the third section, short definition of the fractional derivative in the Caputo sense is introduced. In the fourth section, Bernoulli collocation method is explained and adopted to the present problem. In the fifth section, obtained results are given and discussions are made in the light of employing the Bernoulli collocation method to fractional viscoelastic beam system.

2. Definition of the Problem

The motion equation of the fractional viscoelastic homogeneous beam is obtained by considering the Euler–Bernoulli beam theory by ignoring shear deformation factor and rotary inertia of the beam. The beam is considered as a uniform viscoelastic beam and mechanical energy dissipation inside the beam is modeled by fractional order differential equations. By taking into account the [35], stressstrain constitutive relation of a fractional viscoelastic beam is given as follows:

$$\sigma = E\varepsilon(t) + E_{\gamma}' D_t^{\gamma}[\varepsilon(t)] = E\left(\varepsilon + \mu_{\gamma} \frac{d^{\gamma}\varepsilon(t)}{dt^{\gamma}}\right), \quad (1)$$

in which *E* is the Young's modulus of the viscoelastic beam, μ_{γ} is the damping coefficient, and D_t^{γ} is the fractional derivative operator with the order γ with respect to *t*. The simply supported viscoelastic beam initially is at rest and nondeformed. The beam is subjected to a horizontally moving constant force load with the velocity v(t) from the left edge to right edge of the beam, respect to *x* axis. In the light of [27], let us introduce the formulation of a fractional viscoelastic beam structure illustrated in Figure 1.

$$A\rho \frac{\partial^2 w(t,x)}{\partial t^2} + \mathrm{EI}\mu_{\gamma} \left[\frac{d^{\gamma}}{dt^{\gamma}} \frac{\partial^4 w(t,x)}{\partial x^4} \right] + \mathrm{EI} \frac{\partial^4 w(t,x)}{\partial x^4} = P\delta(x-v(t)), \tag{2}$$

in which w is the deflection of the viscoelastic beam in $\mathscr{C} = \{(t, x): t \in (0, \text{tf}), x \in (0, \ell)\}, t \text{ is the time variable, } t_f \text{ is the final time observed duration, } x \text{ is the space variable, } \ell \text{ is the length of the viscoelastic beam, } A \text{ is the cross-section area of the structure, } \rho \text{ is the material mass density of the viscoelastic beam, } I \text{ is the axial moment of inertia of the beam, } P \text{ is a constant showing intensity of the external moving force load, } \delta \text{ is the Dirac-delta function, and } v(t) \text{ is the velocity of the moving force load with the condition } 0 \le v(t) \le \ell$. Equation (2) is subjected to the following boundary conditions:

$$w(t, x) = 0, w_{xx}(t, x) = 0 \text{ at } x = 0, \ell,$$
 (3)

and the following initial conditions:

$$w(t, x) = w_0(x), w_t(t, x) = w_1(x) \text{ at } t = 0,$$
 (4)

in which
$$w_0(x) \in H^1(0, \ell) = \{w_0(x) \in L^2(0, \ell): \partial w_0(x)/\partial x \in L^2(0, \ell)\}, w_1(x) \in L^2(0, \ell). L^2(\mathscr{C}) \text{ means to square-integrable functions space in the manner of Hilbert in the domain \mathscr{C} in the Lebesgue sense with the following norm and inner product:$$

$$\|\eta\|^2 = \langle \eta, \eta \rangle, \quad \langle \eta, \rho \rangle_{\mathscr{C}} = \int_{\mathscr{C}} \rho \eta \, d\mathscr{C}. \tag{5}$$

Let us assume that

$$w(t,x) = \sum_{n=1}^{N} z_n(t)\sqrt{2}\sin\left(\frac{n\pi x}{\ell}\right).$$
(6)

After substituting the equations (6) into (2) and multiplying both sides of equation (2) with $\sqrt{2} \sin(n\pi x/\ell)$, integrating on $(0, \ell)$, we obtain the following ordinary differential equation as follows:

$$A\rho z_n''(t) + \operatorname{EI}(n\pi)^4 \mu_{\gamma} \left[\frac{d^{\gamma}}{dt^{\gamma}} z_n(t) \right] + \operatorname{EI}(n\pi)^4 z_n(t) = P\sqrt{2} \sin\left(\frac{n\pi\nu(t)}{\ell}\right), \quad n = 1, \dots, N.$$

$$(7)$$

Equation (7) is subjected to the following initial conditions:

$$z_n(0) = \sqrt{2} \int_0^\ell w_0(x) \sin\left(\frac{n\pi x}{\ell}\right) \mathrm{d}x, \quad z_n'(0) = \sqrt{2} \int_0^\ell w_1(x) \sin\left(\frac{n\pi x}{\ell}\right) \mathrm{d}x. \tag{8}$$

3. The Fractional Derivative in the Caputo Sense

Definition. The Caputo definition of the fractional-order derivative is

$$D^{\gamma}f(x) = \frac{1}{\Gamma(n-\gamma)} \int_{0}^{x} \frac{f^{(n)}(t)}{(x-t)^{\gamma+1-n}} dt, \quad n-1 < \gamma \le n, \ n \in \mathbb{N},$$
(9)

where $\gamma > 0$ is the order of the derivative and *n* is the smallest integer greater than γ . For the Caputo derivative, we have



FIGURE 1: Schematic of the viscoelastic beam under moving force load P with the velocity v(t).

 $D^{\gamma}C = 0$, *C* is constant,

$$D^{\gamma} x^{q} = \begin{cases} 0, & \text{for } q \in \mathbb{N}_{0} \text{ and } q < \lceil \gamma \rceil, \\ \frac{\Gamma(q+1)}{\Gamma(q+1-\gamma)} x^{q-\gamma}, & \text{for } q \in \mathbb{N}_{0} \text{ and } q \ge \lceil \gamma \rceil \text{ or } q \notin \mathbb{N} \text{ and } q > \lfloor \gamma \rfloor. \end{cases}$$

$$(10)$$

4. Bernoulli Collocation Method

The recurrence relation of the Bernoulli polynomials is defined by

$$B_n(x) = 2xB_{n-1}(x) + B_{n-2}(x).$$
(11)

For $n \ge 3$, $B_1(x) = 1$, $B_2(x) = 2x$. The first few Bernoulli polynomials are

$$B_1(x) = 1,$$
 (12)

$$B_{2}(x) = x - \frac{1}{2},$$

$$B_{3}(x) = x^{2} - x - \frac{1}{6},$$

$$B_{4}(x) = x^{3} - \frac{3}{2}x^{2} + \frac{x}{2}.$$

:
:
:
(13)

Our goal is to get the approximate solution as the truncated Bernoulli series defined by

$$y(x) = \sum_{n=1}^{N+1} c_n B_n(x),$$
(14)

where $B_n(x)$ denotes the Bernoulli polynomials; $c_n (1 \le n \le N + 1)$ are the unknown coefficients for Bernoulli polynomial, and N is any positive integer which possess $N \ge m$. Let us assume that linear combination of Bernoulli polynomials equation (14) is an approximate solution of equation (7). Our purpose is to determine the matrix forms of equation (7) by using (14). Firstly, we can write Bernoulli polynomials (12) in the matrix form

$$\mathbf{B}(x) = \mathbf{T}(x)\mathbf{M},\tag{15}$$

where $B(x) = [B_1(x) B_2(x) \cdots B_{N+1}(x)], \quad \mathbf{T}(x) = (1 x x^2 x^3 \dots x^N), \mathbf{C} = (c_1 c_2 \cdots c_{N+1})^T$, and

	(1	$\frac{1}{2}$	$\frac{1}{6}$	0	$-\frac{1}{30}$	0	$\frac{1}{42}$	0	$-\frac{1}{30}$	
	0	1	-1	$\frac{1}{2}$	0	$\frac{1}{6}$	0	$\frac{1}{6}$	0	
	0	0	1	$-\frac{3}{2}$	1	0	$-\frac{1}{2}$	0	$\frac{2}{3}$	
	0	0	0	1	-2	$\frac{5}{3}$	0	$-\frac{7}{6}$	0	
M =	0	0	0	0	1	$-\frac{5}{2}$	$\frac{5}{2}$	0	$-\frac{7}{3}$	(16)
	0	0	0	0	0	1	-3	$\frac{7}{2}$	0	
	0	0	0	0	0	0	1	$-\frac{7}{2}$	$\frac{14}{3}$	
	0	0	0	0	0	0	0	1	-4	
	0	0	0	0	0	0	0	0	1	

The matrix form of equation (14) by a truncated Bernoulli series is given by

$$y(x) = \mathbf{B}(x)\mathbf{C}.$$
 (17)

By using equations (15) and (17), the matrix relation is expressed as

$$y(x) \approx y_N(x) = \mathbf{T}(x)\mathbf{MC},$$

$$y^{(\gamma)}(x) \approx y_N^{(\gamma)}(x) = \mathbf{T}(x)\mathbf{X}_{(\gamma)}(x)\mathbf{D}_{(\gamma)}\mathbf{MC},$$
(18)

$$y''(x) \approx y_N''(x) = \mathbf{T}(x)\mathbf{D}^2\mathbf{MC},$$

D

where

$$\mathbf{X}_{(\gamma)}(x) = \left[0, x^{1-\gamma}, x^{2-\gamma}, \dots, x^{N-\gamma}\right],$$
(19)

$$\mathbf{D} = \begin{bmatrix} 0 & 1 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 2 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 3 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 4 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & \cdots & 0 \\ 0 & \frac{\Gamma(2)}{\Gamma(2 - \gamma)} & 0 & \cdots & 0 \\ 0 & 0 & \frac{\Gamma(3)}{\Gamma(3 - \gamma)} & \cdots & 0 \\ \vdots & \vdots & \vdots & \vdots & \ddots & \vdots \\ 0 & 0 & 0 & 0 & \cdots & \frac{\Gamma(N)}{\Gamma(N - \gamma)} \end{bmatrix}.$$
(20)

By using equation (18), we obtain the following relation:

$$\mathbf{Y}^{(k)}(x) = \mathbf{T}(x)\mathbf{D}^{k}\mathbf{M}\mathbf{C}.$$
 (21)

By substituting the Bernoulli collocation points given by

$$x_i = a + \frac{(b-a)i}{N}, \quad i = 0, 1, \dots, N,$$
 (22)

into equation (21), we obtain

$$\mathbf{Y}^{(k)}(x_i) = \mathbf{T}(x_i)\mathbf{D}^k\mathbf{M}\mathbf{C}, \quad k = 0, \gamma, 2.$$
(23)

and the compact form of the relation (23) becomes

$$\mathbf{Y}^{(k)} = \mathbf{TD}^{k}\mathbf{MC}, \quad k = 0, \gamma, 2.$$
 (24)

In this way, the unknown Bernoulli coefficients c_n , n = 1, 2, ..., N + 1 are obtained by solving the system. Then, these coefficients are substituted into (14), and the approximate solution is obtained. For more details, see [36].

5. Simulation Results and Discussion

Bernoulli collocation method for obtaining the solution of fractional viscoelastic beam equation is employed. Hence, displacement analysis of a forced fractional viscoelastic beam is investigated by taking into account the different moving force loads, different values of internal damping coefficient, and different values of fractional order of derivative. Obtained results are simulated and presented in the tables and

graphical forms. The velocity, from left to right, of the external moving force on beam v(t) is considered as $\sin(\pi t)$. In order to observe the dynamic response of the viscoelastic beams under the different intensity of external moving force, the intensity constant of the external moving force load on the beam is involved to computation as P = 1, 25, 50. Also, the values in Tables 1–6 are computed on x = 0.5, which is the middle point of the fractional viscoelastic beams. Observed duration of time is $t_f = 1$. In the first case, forced displacement analysis of a polybutadiene beam is observed for different values of moving force load and results are presented in Table 1. The length and material density of the fractional viscoelastic beam are taken into account as $\ell = 1 m$ and $\rho = 160 \text{ kg/m}^3$, respectively. The cross-sectional area A is 0.72 m^2 , moment of inertia J is $(0.1)^4/12$, and Young's modulus E is 8.15×10^5 for a fractional viscoelastic polybutadiene beam. Also, the order of fractional derivative γ is evaluated as 0.528 for the results in Figure 2 and Tables 1 and 3. By observing Figure 2, it is concluded that while the intensity of the external moving load force increases, namely, P is 1 to 25 and 50, the displacement of the fractional viscoelastic polybutadiene beam also increases. Also, parallel observation results to Figure 2 are obtained by taking into account Table 1. For example, on the moment t = 0.5, the amount of the displacement of the polybutadiene beam is measured as 0.001 1 for P = 1, 0.028 for P = 25, and 0.056 for P = 50. This observation is valid the entire time interval t = 0, ..., 1 for polybutadiene beam. Also, the effect of internal damping on the displacement is presented in Table 3 for polybutadiene beam. The internal damping

t	$w_{P=1}$	$w_{P=25}$	$w_{P=50}$
0.1	0.000 025 2	0.000 632 0	0.001 264 0
0.2	0.0001761	0.004 402 5	0.008 805 0
0.3	0.000 463 7	0.011 593 4	0.023 186 8
0.4	0.000 805 8	0.020 145 2	0.040 290 4
0.5	0.001 115 0	0.027 877 2	0.0557545
0.6	0.001 350 9	0.033 774 5	0.067 549 0
0.7	0.001 529 4	0.038 236 1	0.076 472 1
0.8	0.001 709 4	0.042 735 6	0.085 471 2
0.9	0.001 933 0	0.048 326 2	0.096 652 5
1.0	0.002 026 3	0.050 657 4	0.101 315 0

TABLE 2: Some values of w(t, x) for P = 1, 25, 50 (for a butyl B252 beam).

t	$w_{P=1}$	$w_{P=25}$	$w_{P=50}$
0.1	0.000 020 2	0.000 505 3	0.001 010 7
0.2	0.0001407	0.003 517 7	0.0070354
0.3	0.000 370 1	0.009 253 1	0.018 506 4
0.4	0.000 642 1	0.016 053 0	0.0321061
0.5	0.000 886 5	0.022 164 5	0.044 328 9
0.6	0.001 070 9	0.026 772 4	0.053 544 8
0.7	0.001 207 9	0.030 198 1	0.060 396 2
0.8	0.001 345 0	0.033 624 9	0.067 249 8
0.9	0.001 515 9	0.037 899 2	0.075 798 4
1.0	0.001 581 7	0.039 544 7	0.079 089 3

TABLE 3: Some values of w(t, x) for different values of μ for P = 1 (for a polybutadiene beam).

tμ	0.2	0.4	0.6	0.8	1.0
0.1	0.000 024 90	0.000 024 24	0.000 023 62	0.000 023 03	0.000 022 48
0.2	0.000 170 50	0.00016104	0.000 152 42	0.000 144 54	0.000 137 32
0.3	0.00044042	0.000 402 52	0.000 369 54	0.000 340 70	0.000 315 37
0.4	0.000 748 72	0.000 659 75	0.000 586 20	0.000 524 91	0.00047341
0.5	0.001 011 32	0.000 856 73	0.000 735 83	0.000 640 02	0.000 563 11
0.6	0.001 194 99	0.000 973 52	0.000 810 05	0.000 687 00	0.000 592 56
0.7	0.001 323 26	0.001 044 56	0.000 850 35	0.000 711 06	0.000 608 30
0.8	0.001 459 32	0.001 136 46	0.000 922 48	0.000 774 52	0.000 667 97
0.9	0.001 645 41	0.001 287 75	0.001 058 63	0.000 902 75	0.000 790 71
1.0	0.001 713 62	0.001 340 71	0.001 112 91	0.000 964 62	0.000 862 85

TABLE 4: Some values of w(t, x) for different values of μ for P = 1 (for a butyl B252 beam).

tμ	0.2	0.4	0.6	0.8	1.0
0.1	0.000 019 90	0.000 019 35	0.000 018 85	0.000 018 37	0.000 017 91
0.2	0.00013609	0.000 128 33	0.000 121 26	0.00011482	0.000 108 94
0.3	0.000 350 96	0.000 319 92	0.000 293 02	0.000 269 58	0.00024907
0.4	0.000 595 32	0.000 522 71	0.000 463 02	0.000 413 52	0.000 372 12
0.5	0.000 801 77	0.000 676 17	0.000 578 62	0.000 501 79	0.00044045
0.6	0.000 943 91	0.000 764 93	0.000 633 94	0.000 536 04	0.000 461 36
0.7	0.001 040 91	0.000 817 11	0.000 662 68	0.000 552 79	0.00047222
0.8	0.001 459 32	0.000 886 19	0.000 717 38	0.000 601 55	0.000 667 97
0.9	0.001 285 91	0.001 002 83	0.000 823 34	0.000 701 97	0.000 518 57
1.0	0.001 333 74	0.001 041 33	0.000 864 75	0.000 750 70	0.000 615 00

coefficient is evaluated from 0.2 to 1 and by examining Table 3; it reveals that when internal damping coefficient increases, the displacement of the polybutadiene beam decreases under the same conditions. The effect of the fractional order to system is observed from Table 5 and it can be concluded that while increasing the values of the

tγ 0.1

0.2

0.3

0.4

0.5

0.6

0.7

0.8

0.9

1.0

0.2

0.000 025 28

0.00017611

0.000 463 79

0.000 805 96

0.001 115 39

0.001 351 46

0.001 530 10

0.001 710 24

0.001 933 99

0.002 027 31

different values of γ for $P = 1$ (for	or a polybutadiene beam).	
0.6	0.8	1.0
0.000 025 28	0.000 025 27	0.000 025 27

0.00017607

0.000 463 63

 $0.000\,805\,56$

0.001 114 65

0.001 350 34

0.001 528 64

0.001 708 52

0.001 932 09

0.002 025 38

TABLE 5: Some values of w(t, x) for

0.00017609

0.000 463 71

0.000 805 75

0.001 114 99

 $0.001\,350\,84$

0.001 529 25

0.001 709 20

0.001 932 81

0.002 026 06

0.4

 $0.000\,025\,28$

0.00017610

 $0.000\,463\,76$

 $0.000\,805\,88$

0.001 115 23

0.001 351 20

0.001 529 73

0.001 709 78

0.001 933 45

0.002 026 71

TABLE 6: Some values of w(t, x) for different values of γ for P = 1 (for a butyl B252 beam).

tγ	0.2	0.4	0.6	0.8	1.0
0.1	0.000 020 21	0.000 020 21	0.000 020 21	0.000 020 21	0.000 020 21
0.2	0.000 140 71	0.00014071	0.000 140 70	0.000 140 69	0.00014066
0.3	0.000 370 16	0.000 370 14	0.000 370 11	0.000 370 06	0.000 369 97
0.4	0.000 642 21	0.000 642 16	0.000 642 08	0.000 641 96	0.000 641 77
0.5	0.000 886 76	0.000 886 66	0.000 886 51	0.000 886 29	0.000 885 98
0.6	0.001 071 19	0.001 071 02	0.001 070 79	0.001 070 48	0.00107007
0.7	0.001 208 33	0.001 208 09	0.001 207 79	0.001 207 41	0.001 206 93
0.8	0.001 345 49	0.001 345 20	0.001 344 84	0.001 344 41	0.001 343 92
0.9	0.001 516 53	0.001 516 20	0.001 515 80	0.001 515 36	0.001 514 89
1.0	0.001 582 39	0.001 582 02	0.001 581 62	0.001 581 21	0.001 580 82



FIGURE 2: Continued.

0.00017604 $0.000\,463\,49$

 $0.000\,805\,26$

 $0.001\,114\,16$

0.001 349 69

 $0.001\ 527\ 88$ $0.001\,707\,72$

0.00079071

0.001 931 32



FIGURE 2: Displacements of a polybutadiene beam for P = 1, 25, 50.



FIGURE 3: Displacements of a butyl B252 beam for P = 1, 25, 50.

fractional derivative, the value of the displacement is decreasing. In the second case, a butyl B252 beam is taken into account by the coefficients; the cross-sectional area A is 0.72 m^2 , moment of inertia J is $(0.1)^4/12$, and Young's modulus E is 1.05×10^6 . The order of fractional derivative y is considered as 0.519 for Figure 3 and Tables 2 and 4. By checking Figure 3, it is easy to see that displacements corresponding to much bigger intensity of moving force load are much bigger. For example, on the moment t = 0.5, while P = 1 to P = 25, 50, corresponding displacements are calculated as 0.00089, 0.022, and 0.044, respectively. This is effective along the observation duration. In Table 4, some results related to the effect of internal damping are presented and internal damping coefficient is included in the computation as 0.2 to 1. After looking at Table 4, it is concluded that while internal damping coefficient decreases, the displacement of the butyl B252 beam increases and relation between the effects of internal damping and displacements is inversely proportional. The relation between the displacement and fractional order in the system is vice versa. As understood from Table 6, while decreasing the values of the fractional derivative, the value of the displacement is increasing. These observation results of the present study are also compatible with the results existing in the literature. By taking into account Tables 1-6 and Figures 2 and 3 and comparing these two kinds of fractional viscoelastic beams, it is seen that the polybutadiene beam has more greater displacements than butyl B252 beam under same conditions. Also, the effect of internal damping coefficient is more visible on the butyl B252 beam according to polybutadiene beam. These observations make clear that butyl B252 beam is stronger and preferable than the polybutadiene beam.

6. Conclusion

In this study, the Bernoulli collocation method as a new solution method for obtaining the approximate solution of a fractional viscoelastic beam model subjected to moving force load is employed. Dynamic response analysis of the fractional viscoelastic beam model is investigated for two different specific beams: polybutadiene beam and butyl B252 beam. Displacement analysis of a point on the fractional viscoelastic beams is studied for different moving force loads and also effect of the internal damping to displacement is observed for different internal damping coefficients. Moreover, dynamic response of the fractional viscoelastic beam is examined for different values of the fractional order. Obtained results are presented in tables and graphics and results reveal that Bernoulli collocation method is very effective and powerful solution method for obtaining the solution of fractional order viscoelastic beam models. After observing Figures 2 and 3, it is easy to conclude that as the moving force load increases, the displacement of a point on the beams also increases. Also, numerical results, presented in Tables 1-4, show that under the same moving force load with the same internal damping effect, the displacement of a point on the polybutadiene beam is greater than that corresponding to butyl B252 beam. Moreover, under the same

moving force load, changes in the displacements of a point on the beams are examined in the aspect of different internal damping effects and observations made clear that butyl B252 beam better reflects the effect of internal damping to displacement of a point on the beam. By comparing polybutadiene beam and butyl B252 beam, it is concluded that polybutadiene beam is more open to destructive effects of vibrations under the same conditions with the butyl B252 beam.

Data Availability

The data used to support the findings of this study are included within the article.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

Authors' Contributions

The authors completed this study and wrote and approved the final version of the manuscript.

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Research Article

Several Characterizations on Degree-Based Topological Indices for Star of David Network

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In order to make quantitative structure-movement/property/danger relations, topological indices (TIs) are the numbers that are related to subatomic graphs. Some fundamental physicochemical properties of chemical compounds, such as breaking point, protection, and strain vitality, correspond to these TIs. In the compound graph hypothesis, the concept of TIs was developed in view of the degree of vertices. In investigating minimizing exercises of Star of David, these indices are useful. In this study, we explore the different types of Zagreb indices, Randić indices, atom-bond connectivity indices, redefined Zagreb indices, and geometric-arithmetic index for the Star of David. The edge partitions of this network are tabled based on the sum of degrees-of-end vertices and the sum of degree-based edges. To produce closed formulas for some degree-based network TIs, these edge partitions are employed.

1. Introduction

Graph theory is a branch of mathematics in which we use graph parameter methods to precisely expose the compound phenomenon. For example, the graph theory characterizes an area between different disciplines of science when applied to the investigation of molecular structures, which is known as molecular topology or the theory of chemical graphs. A significant part of the analysis was supported by chemical graph theory [1]. Chemist can be performed for the statistical demonstration of chemical marvel by means of graph theory. In quantitative structure activity, researchers tried to figure out what structural characteristics will be developed. Physicochemical features and topological measures are discussed by Wiener. Different types of graph descriptors, such as distance-based, degree-based, spectral, and polynomial-related descriptors, have been well defined and explored extensively in the literature. Vertex degree-based descriptors are the most important of these classes, and they

play a crucial role in chemical graph theory. These descriptors are combined to infer physicochemical, biological, and pharmacological qualities such stability, chirality, melting point, boiling point, similarity, connectedness, entropy, enthalpy of formation, surface tension, density, critical temperature, and others. Mathematicians and chemists use a variety of topological indices in these types of studies. The quantitative structure-property relationship (QSPR) and quantitative structure-activity relationship (QSAR) research use the index, the Randic index, the Zagreb indices, and the ABC index to measure by Yang et al. [2] in the bioactivity of chemical compounds [3]. Topological indices provide numerical representations, molecular size, shape, branching, and other properties that are used to compare chemical compounds' topological similarities and in QSPR/QSAR research [4, 5]. There are several properties related to new families of graphs that are discussed in [6-8] such as metric dimensions and indices. The spectral properties, metric dimensions, and indices of different families of graphs are discussed in [9–14]. These include distance-based and degree-based TIs, as well as related polynomials [15] and classified graph indices, among other forms of topological indices. In 2017, Maji and Ghorai introduced new distancedegree-based topological indices, see [16]. In chemical graph theory and notably in chemistry, degree-based TIs are extremely important and serve a critical function. Furthermore, algebraic graph theory results are discussed in [17–21] by using the notion of totient number which was introduced by Shahbaz and Khalid in 2017.

In this paper, we study some degree-based analysis of TIs of the Star of David network. In Hebrew, the Star of David, or "Magen David" ("Shield of David"), consists of two overlaid equilateral triangles that form a six-pointed star. It cannot be traced back to the Bible or the Talmud, but it is said that it comes from the (presumed) similarity to the shape of the shield of King David. It was neither used as a sign of Jewish identity, although it originated in Antiquity, nor was even restricted to Judaism. The seven-branched candelabrum, still one of Israel's emblems today, was the most famous symbol of Judaism at that time.

We arranged our paper as follows. In Section 2, we give some preliminary concepts related to topological indices of different kinds. In Section 3, we construct the Star of David network and proposed their algorithm. In Section 4, we compute several results on topological indices for the proposed network. In Section 5, we give a comparison of topological indices for proposed networks. In Section 6, we give the concluding remarks about our proposed work. In future work, one can compute more indices on proposed Star of David networks.

2. Preliminaries

According to this study, a simple connected graph *G* is made up of vertices V(G) and edges E(G), with $\aleph(\mu)$ being the degree of each vertex and the number of edges intersecting μ ."

Definition 1. Topological index (TI) is derived by Wiener, which was created in 1945 after investigating alkane's boiling point [22]. According to Randić [23] characterization, the earliest degree-based index is the Randić index, which is defined as

$$R(G) = \sum_{\mu\nu\in E(G)} \frac{1}{(\sqrt{\aleph(\mu)\aleph(\nu)})}.$$
 (1)

As a traditional graph-based molecular structure descriptor, the Randić index has been widely used in chemical and pharmaceutical research. Even the mathematical sense of this index is clear for detail of its QSPR/QSAR application, see [24, 25]. Bollobác and Paul [26] presented the general Randić index, which is defined as

$$R_{\alpha}(G) = \sum_{\mu\nu \in E(G)} (\aleph(\mu)\aleph(\nu))^{\alpha} \text{ for } \alpha = 1, \frac{1}{2}, -\frac{1}{2}, -1.$$
(2)

Definition 2. Gutman and Trinajstić [27] defined the first and second Zagreb indices as follows; also, see [28–30]:

$$M_{1}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu) + \aleph(\nu)),$$

$$M_{2}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu)\aleph(\mu)).$$
(3)

Definition 3. Ranjini et al. [31] proposed the redefined version of Zagreb indices

The redefined first Zagreb index of graph G is defined as

$$R_e ZG_1(G) = \sum_{\mu\nu \in E(G)} \frac{(\aleph(\mu) + \aleph(\nu))}{(\aleph(\mu) \cdot \aleph(\nu))}.$$
 (4)

The redefined second Zagreb index of graph *G* is defined as

$$R_e ZG_2(G) = \sum_{\mu\nu \in E(G)} \frac{(\aleph(\mu) \cdot \aleph(\nu))}{(\aleph(\mu) + \aleph(\nu))}.$$
(5)

The redefined third Zagreb index of graph G is defined as

$$R_e ZG_3(G) = \sum_{\mu\nu \in E(G)} (\aleph(\mu) \cdot \aleph(\nu)) (\aleph(\mu) + \aleph(\nu)).$$
(6)

Definition 4. Estrada et al., in [32], proposed degree-based TI *ABC* and defined it as

$$ABC(G) = \sum_{\mu\nu\in E(G)} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}}.$$
 (7)

The atom-bond connectivity index (*ABC*) is a molecular structural descriptor that has lately found surprising applicability in explaining linear and branched alkane stability, as well as cycloalkane strain energy. It is used for modeling the thermodynamic characteristics in organic chemical molecules.

Definition 5. The *GA* index is proposed by Vukicevic and Furtula in [33] and defined as

$$GA(G) = \sum_{\mu\nu\in E(G)} 2\frac{\sqrt{\aleph(\mu)}\cdot\aleph(\nu)}{\aleph(\mu)+\aleph(\nu)}.$$
(8)

The prediction value of the *GA* index is slightly greater than the Randić connection index for physicochemical characteristics such as entropy and acentric factor, according to [33].

Definition 6. Furtula and Gutman, in [34], proposed the forgotten TI and stated it as

$$F(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu))^2 + (\aleph(\nu))^2.$$
(9)

For further study, see [35]. The Star of David networks are shown in Figures 1 and 2.

3. Higher Dimension SD_(n) **Drawing** Algorithm for Star of David Network

- Step 1 : the Star of David, consists of two equilateral overlapping triangles forming a six-pointed star. Draw David's Star graph *G*, which is one and two dimensional, as seen in figure ??. Algorithm for Star of David is as given below:
 - (i) *≠inclu de*⟨*iostream*⟩

```
using namespace std;, \neq define n \leq n'
      int i, j;, int main()
{
      for (i = 0; i \le 1; i + +)
      {
         for (j = 1; j \le 5 - i; j + +)
   cout \langle \langle " " ;
</" * ";
if (i == 1)
   cout \langle \langle " * ";
   cout \langle \langle n;
      }
      for (i = 1; i \le ; i + +)
         {
      for (j = 1; j \le 11; j + +)
           {
if (i == 2 || i == 3)
if (j == 2 || j == 3 || j == 9 || j == 10)
   \cot \langle \langle " * ";
   }
   else
            cout \langle \langle " " \rangle;
```

- Step 2 : add two David's stars on the upper and lower sides.
- Step 3 : adding one more star of David in each step similarly, we proceed up to n = 5. We get the sequence at n = 5.

4. Results on Indices for Star of David Network

Theorem 1. The atom-bond connectivity index of Star of David network is

$$ABC(G) = 3.2n^2 - 10.6n - 10.2.$$
(10)

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the atom-bond connectivity index for *G*:

$$ABC(G) = \sum_{\mu\nu\in E(G)} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} \\ = \sum_{\mu\nu\in E\{2,2\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} + \sum_{\mu\nu\in E\{2,4\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} + \sum_{\mu\nu\in E\{2,4\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} \\ + \sum_{\mu\nu\in E\{4,4\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} + \sum_{\mu\nu\in E\{4,6\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} + \sum_{\mu\nu\in E\{6,6\}} \sqrt{\frac{\aleph(\mu) + \aleph(\nu) - 2}{\aleph(\mu) \cdot \aleph(\nu)}} \\ = 4\sqrt{\frac{2 + 2 - 2}{2 \cdot 2}} + 8\sqrt{\frac{2 + 4 - 2}{2 \cdot 4}} + (12n - 16)\sqrt{\frac{2 + 6 - 2}{2 \cdot 6}} \\ + (6n - 6)\sqrt{\frac{4 + 4 - 2}{4 \cdot 4}} + (12n - 20)\sqrt{\frac{4 + 6 - 2}{4 \cdot 6}} + 6(n - 2)^2\sqrt{\frac{6 + 6 - 2}{6 \cdot 6}}.$$
(11)





FIGURE 2: Star of David network.

TABLE 1: Star of David network edge partition.

Types of edges	$E_{\{2,2\}}$	$E_{\{2, 4\}}$	$E_{\{2, 6\}}$	$E_{\{4,4\}}$	$E_{\{4,6\}}$	$E_{\{6,6\}}$
Number of edges	(2,2)	(2, 4)	(2,6)	(4, 4)	(4,6)	(6,6)
Frequency	4	8	12 <i>n</i> – 16	6n - 6	12n - 20	$6(n-2)^2$

$$ABC(G) = 3.2n^2 - 10.6n - 10.2.$$
(12)

Theorem 2. *The geometric-arithmetic index of Star of David network is*

$$GA(G) = 6n^{2} + 4.2n - 6.46.$$
(13)

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of geometric-arithmetic index for *G*:

$$GA(G) = \sum_{\mu\nu\in E(G)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)}$$
$$= \sum_{\mu\nu\in E(2,2)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)} + \sum_{\mu\nu\in E(2,4)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)} + \sum_{\mu\nu\in E(2,6)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)}$$

$$+ \sum_{\mu\nu\in E(4,4)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)} + \sum_{\mu\nu\in E(4,6)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)} + \sum_{\mu\nu\in E(6,6)} 2\frac{\sqrt{\aleph(\mu)\cdot\aleph(\nu)}}{\aleph(\mu)+\aleph(\nu)}$$

$$= 4 \cdot 2\frac{\sqrt{2\cdot2}}{2+2} + 8 \cdot 2\frac{\sqrt{2\cdot4}}{2+4} + (12n-16) \cdot 2\frac{\sqrt{2\cdot6}}{2+6}$$

$$+ (6n-6) \cdot 2\frac{\sqrt{4\cdot4}}{4+4} + (12n-20) \cdot 2\frac{\sqrt{4\cdot6}}{4+6} + 6(n-2)^2 \cdot 2\frac{\sqrt{6\cdot6}}{6+6}$$

$$= 6n^2 + 4.2n - 6.46.$$

$$(14)$$

Theorem 3. The first Zagreb index of Star of David network is

$$M_1(G) = 72n^2 - 24n - 24. \tag{15}$$

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the first Zagreb index for *G*:

$$M_{1}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu) + \aleph(\nu))$$

$$= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu) + \aleph(\nu))$$

$$+ \sum_{\mu\nu\in E(4,4)} (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu) + \aleph(\nu))$$

$$= 4(2+2) + 8(2+4) + (12n - 16)(2+6) + (6n - 6)(4+4)$$

$$+ (12n - 20)(4+6) + 6(n - 2)^{2}(6+6).$$
(16)

We get the outcomes after estimates:

$$M_1(G) = 72n^2 - 24n - 24.$$
(17)

Theorem 4. The second Zagreb index of Star of David network is

2

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the second Zagreb index for *G*:

 $M_2(G) = 216n^2 - 336n + 176.$

$$M_{2}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu) \cdot \aleph(\nu))$$

$$= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu) \cdot \aleph(\nu))$$

$$+ \sum_{\mu\nu\in E(4,4)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu) \cdot \aleph(\nu))$$

$$= 4(2 \cdot 2) + 8(2 \cdot 4) + (12n - 16)(2 \cdot 6) + (6n - 6)(4 \cdot 4) + (12n - 20)(4 \cdot 6)$$

$$+ 6(n - 2)^{2}(6 \cdot 6)$$

$$= 216n^{2} - 336n + 176.$$
(19)

Theorem 5. *The redefined first Zagreb index of Star of David network is*

$$R_e ZG_1(G) = 2n^2 + 8n - 3.97.$$
(20)

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the redefined first Zagreb index for *G*:

(18)

$$R_{e}ZG_{1}(G) = \sum_{\mu\nu\in E(G)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)}$$

$$= \sum_{\mu\nu\in E(2,2)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)} + \sum_{\mu\nu\in E(2,4)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)} + \sum_{\mu\nu\in E(2,6)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)}$$

$$+ \sum_{\mu\nu\in E(4,4)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)} + \sum_{\mu\nu\in E(4,6)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)} + \sum_{\mu\nu\in E(6,6)} \frac{\aleph(\mu) + \aleph(\nu)}{\aleph(\mu) \cdot \aleph(\nu)}$$

$$= 4 \cdot \frac{2+2}{2\cdot 2} + 8 \cdot \frac{2+4}{2\cdot 4} + (12n - 16) \cdot \frac{2+6}{2\cdot 6} + (6n - 6) \frac{4+4}{4\cdot 4}$$

$$+ (12n - 20) \frac{4+6}{4\cdot 6} + 6(n - 2)^{2} \frac{6+6}{6\cdot 6}.$$
(21)

$$R_e ZG_1(G) = 2n^2 + 8n - 3.97.$$
(22)

Theorem 6. The redefined second Zagreb index of Star of David network is

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the redefined second Zagreb index for *G*:

$$R_{e}ZG_{2}(G) = \sum_{\mu\nu\in E(G)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)}$$

$$= \sum_{\mu\nu\in E(2,2)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)} + \sum_{\mu\nu\in E(2,4)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)} + \sum_{\mu\nu\in E(2,6)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)}$$

$$+ \sum_{\mu\nu\in E(4,4)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)} + \sum_{\mu\nu\in E(4,6)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)} + \sum_{\mu\nu\in E(6,6)} \frac{\aleph(\mu) \cdot \aleph(\nu)}{\aleph(\mu) + \aleph(\nu)}$$

$$= 4 \cdot \frac{2 \cdot 2}{2 + 2} + 8 \cdot \frac{2 \cdot 4}{2 + 4} + (12n - 16) \cdot \frac{2 \cdot 6}{2 + 6} + (6n - 6) \frac{4 \cdot 4}{4 + 4}$$

$$+ (12n - 20) \frac{4 \cdot 6}{4 + 6} + 6(n - 2)^{2} \frac{6 \cdot 6}{6 + 6}.$$

$$R_{e}ZG_{3}(G) = 2592n^{2} - 5568n + 3712.$$
(26)

We get the outcomes after estimates:

$$R_e ZG_2(G) = 18n^2 - 13.2n + 2.67.$$
 (25)

Theorem 7. The redefined third Zagreb index of Star of David network is

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the redefined third Zagreb index for *G*:

$$\begin{split} R_e ZG_3(G) &= \sum_{\mu\nu \in E(G)} \left(\aleph(\mu) \cdot \aleph(\nu)\right) \left(\aleph(\mu) + \aleph(\nu)\right) \\ &= \sum_{\mu\nu \in E(2,2)} \left(\aleph(\mu) \cdot \aleph(\nu)\right) \left(\aleph(\mu) + \aleph(\nu)\right) + \sum_{\mu\nu \in E(2,4)} \left(\aleph(\mu) \cdot \aleph(\nu)\right) \left(\aleph(\mu) + \aleph(\nu)\right) \end{split}$$

$$+\sum_{\mu\nu\in E(2,6)} (\aleph(\mu) \cdot \aleph(\nu)) (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(4,4)} (\aleph(\mu) \cdot \aleph(\nu)) (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu) \cdot \aleph(\nu)) (\aleph(\mu) + \aleph(\nu)) + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu) \cdot \aleph(\nu)) (\aleph(\mu) + \aleph(\nu)) = 4(2 \cdot 2) (2 + 2) + 8(2 \cdot 4) (2 + 4) + (12n - 16) (2 \cdot 6) (2 + 6) + (6n - 6) (4 \cdot 4) (4 + 4) + (12n - 20) (4 \cdot 6) (4 + 6) + 6(n - 2)^{2} (6 \cdot 6) (6 + 6).$$
(27)

$$R_e ZG_3(G) = 2592n^2 - 5568n + 3712.$$
 (28)

Theorem 8. The forgotten TI of Star of David network is

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of forgotten TI for *G*:

 $F(G) = 432n^2 - 432n + 48.$

$$F(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2}$$

$$= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2} + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2} + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2}$$

$$+ \sum_{\mu\nu\in E(4,4)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2} + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2} + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu))^{2} + (\aleph(\nu))^{2}$$

$$= 4(2^{2} + 2^{2}) + 8(2^{2} + 4^{2}) + (12n - 16)(2^{2} + 6^{2}) + (6n - 6)(4^{2} + 4^{2})$$

$$+ (12n - 20)(4^{2} + 6^{2}) + 6(n - 2)^{2}(6^{2} + 6^{2}).$$
(30)

We get the outcomes after estimates:

Theorem 9. The Randić indices of Star of David network are $R_1(G) = 216n^2 - 336n + 176,$

 $R_{1/2}(G) = 36n^2 + 220.4n + 85.4,$

 $R_{-1/2}(G) = n^2 + 3.4n - 1.4,$

$$F(G) = 432n^2 - 432n + 48.$$
(31)

Proof. Let *G* be the graph of Star of David network. By using Table 1, we apply the formula of the general Randić index for *G*:

$$R_{\alpha}(G) = \sum_{\mu\nu \in E(G)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{\alpha}.$$
(33)

For $\alpha = 1$,

$$\begin{split} R_{-1}(G) &= 0.17n^2 + 1.2n - 0.54. \\ R_1(G) &= \sum_{\mu\nu\in E(G)} (\aleph(\mu) \cdot \aleph(\nu)) \\ &= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu) \cdot \aleph(\nu)) \\ &+ \sum_{\nu\in G} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\nu\in G} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\nu\in G} (\aleph(\mu) \cdot \aleph(\nu)) + \sum_{\nu\in G} (\aleph(\mu) \cdot \aleph(\nu)) \end{split}$$

(32)

$$\mu_{\nu \in E(4,4)} = 4(2 \cdot 2) + 8(2 \cdot 4) + (12n - 16)(2 \cdot 6) + (6n - 6)(4 \cdot 4) + (12n - 20)(4 \cdot 6) + 6(n - 2)^{2}(6 \cdot 6).$$

(29)

(34)

$$R_1(G) = 216n^2 - 336n + 176.$$
(35)

$$R_{1/2}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2}$$

$$= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2} + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2} + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2}$$

$$\sum_{\mu\nu\in E(4,4)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2} + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2} + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu) \cdot \aleph(\nu))^{1/2}$$

$$= 4(2 \cdot 2)^{1/2} + 8(2 \cdot 4)^{1/2} + (12n - 16)(2 \cdot 6)^{1/2} + (6n - 6)(4 \cdot 4)^{1/2}$$

$$+ (12n - 20)(4 \cdot 6)^{1/2} + 6(n - 2)^2(6 \cdot 6)^{1/2}.$$
(36)

For $\alpha = 1/2$,

We get the outcomes after estimates:

For
$$\alpha = -1/2$$
,

$$R_{1/2}(G) = 36n^2 + 220.4n + 85.4.$$
(37)

$$R_{-1/2}(G) = \sum_{\mu\nu\in E(G)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2}$$

$$= \sum_{\mu\nu\in E(2,2)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2} + \sum_{\mu\nu\in E(2,4)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2} + \sum_{\mu\nu\in E(2,6)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2}$$

$$\sum_{\mu\nu\in E(4,4)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2} + \sum_{\mu\nu\in E(4,6)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2} + \sum_{\mu\nu\in E(6,6)} (\aleph(\mu) \cdot \aleph(\nu))^{-1/2}$$

$$= 4 (2 \cdot 2)^{-1/2} + 8 (2 \cdot 4)^{-1/2} + (12n - 16) (2 \cdot 6)^{-1/2} + (6n - 6) (4 \cdot 4)^{-1/2}$$

$$+ (12n - 20) (4 \cdot 6)^{-1/2} + 6 (n - 2)^{2} (6 \cdot 6)^{-1/2}.$$
(38)

We get the outcomes after estimates:

$$R_{-1/2}(G) = n^2 + 3.4n - 1.4.$$
(39)

For
$$\alpha = -1$$
,

$$\begin{aligned} R_{-1}(G) &= \sum_{\mu\nu\in E(G)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} \\ &= \sum_{\mu\nu\in E(2,2)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} + \sum_{\mu\nu\in E(2,4)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} + \sum_{\mu\nu\in E(2,6)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} \\ &+ \sum_{\mu\nu\in E(4,4)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} + \sum_{\mu\nu\in E(4,6)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} + \sum_{\mu\nu\in E(6,6)} \left(\aleph(\mu) \cdot \aleph(\nu)\right)^{-1} \\ &= 4\left(2 \cdot 2\right)^{-1} + 8\left(2 \cdot 4\right)^{-1} + (12n - 16)\left(2 \cdot 6\right)^{-1} + (6n - 6)\left(4 \cdot 4\right)^{-1} \\ &+ (12n - 20)\left(4 \cdot 6\right)^{-1} + 6\left(n - 2\right)^{2}\left(6 \cdot 6\right)^{-1}. \end{aligned}$$

$$(40)$$

We get the outcomes after estimates:

$$R_{-1}(G) = 0.17n^2 + 1.2n - 0.54.$$
⁽⁴¹⁾

4.1. 3D Graphical Representation of Topological Indices for Star of David Networks. The TIs of the Star of David Network are illustrated graphically in Figure 3. The evolution of TIs along various parameters is portrayed in graphs. Despite the fact



FIGURE 3: Three-dimensional graphical representation of degree-based topological indices.

TABLE 2: Numerical comparison of Star of David network.

п	ABC(G)	GA(G)	$M_1(G)$	$M_2(G)$	$\operatorname{Re}ZG_{1}(G)$	$\operatorname{Re}ZG_{2}(G)$	$\operatorname{Re}ZG_3(G)$	F(G)
1	-17.6	-21.8	24	56	6.03	7.47	736	48
2	-18.6	-149.6	216	368	20.03	48.27	2944	912
3	-13.2	-553.4	552	1112	38.03	125.07	10 336	2640
4	-1.4	-1377.2	1032	2288	60.03	237.87	22 912	5232
5	16.8	-2765	1656	3896	86.03	386.67	40 672	8688
6	41.4	-4860.8	2424	5936	116.03	571.47	63 616	13 008
7	72.4	-7808.6	3336	8408	150.03	792.27	91 744	18192
8	109.8	-11752.4	4392	11 312	188.03	1049.07	125 056	24 2 4 0
9	153.6	-16836.2	5592	14 648	230.03	1341.87	163 552	31 1 52
10	203.8	-23204	6936	18 416	276.03	1670.67	207 232	38 928

TABLE 3: Numerical comparison of Star of David networks.

п	1	2	3	4	5	6	7	8	9	10
$R_1(G)$	56	368	1112	2288	3896	5936	8408	11 312	14 648	18 416
$R_{1/2}(G)$	341.8	670.2	1070.6	1543	2087.4	2703.8	3392.2	4152.6	4985	5889.4
$R_{-1/2}(G)$	3	9.4	17.8	28.2	40.6	55	71.4	89.8	110.2	132.6
$R_{-1}(G)$	0.83	2.54	4.59	6.98	9.71	12.78	16.19	19.94	24.03	28.46



FIGURE 4: Two-dimensional numerical comparison of Star of David network.



FIGURE 5: Two-dimensional numerical comparison of Star of David networks.

that the graphs appear to be similar, their gradients differ. In Figures 4 and 5, we give a two-dimensional numerical comparison of Star of David networks. We discuss the numerical comparison of Star of David networks with different degree-based topological indices in Tables 2 and 3.

5. Comparison of Results for Topological Indices of Star of David Networks

6. Conclusion

TIs for Star of David networks are computed in this paper; as well as, the analytic closed algorithms are reviewed and specified for these networks, namely, the general Randić index, the atomic-bond connectivity index, and the geometric-arithmetic index, as well as the first and second Zagreb index and closed formulas of this network were determined that will help network scientists in understanding and exploring the fundamental topologies of such networks. Computer scientists and chemists who work with Hex-derived networks may find these discoveries valuable.

Data Availability

No data were used to support this study.

Conflicts of Interest

The authors declare no conflicts of interest.

Authors' Contributions

All authors contributed equally to this manuscript. All authors have read and agreed to the published version of the manuscript.

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Research Article On *q*-ANALOGUE of Differential Subordination Associated with Lemniscate of Bernoulli

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This article comprises the study of differential subordination with analogue of q-derivative. It includes the sufficient condition on γ for $1 + (\gamma \partial z_q h(z)/h^n(z))$ to be subordinated by (1 + Az/1 + Bz), $-1 \le B < A \le 1$, and implies that $h(z) < \sqrt{1 + z}$, where h(z) is the analytic function in the open unit disk. Moreover, certain sufficient conditions for q-starlikeness of analytic functions related with lemniscate of Bernoulli are determined.

1. Introduction

Let a set \mathscr{A} be considered as the class of analytic functions defined in open unit disk $\mathbb{U} = \{\varsigma: \varsigma \in \mathbb{C} \text{ and } |\varsigma| < 1\}$ under normalization conditions f(0) = 0 and f'(0) = 1, having

$$f(\varsigma) = \varsigma + \sum_{n=2}^{\infty} a_n \varsigma^n, \quad \varsigma \in \mathbb{U},$$
 (1)

as Taylor series. The class *S* comprises the normalized univalent functions, defined in \mathbb{U} . The major subcategories of class *S* are *C* of convex functions and *S*^{*} of starlike functions. The class *P* is another important class of analytic univalent functions whose co-domains are restricted to the right half plane and are used to determine the convexity and starlikeness of univalent functions. For more details, see [1, 2].

Let *f* and *g* be two analytic functions in \mathbb{U} . Then, *f* is subordinated by *g*, denoted as $f \prec g$ if *f* can be written in the form of composition of *g* and ϖ as $f(\varsigma) = g(\varpi(\varsigma))$ subject to the existence of analytic function ϖ which satisfies the condition that $\varpi(0) = 0$ and $|\varpi(\varsigma)| < |\varsigma|$. Furthermore, if both *f* and *g* are univalent functions in \mathbb{U} , then $f \prec g$ implies that f(0) = g(0) and $f(\mathbb{U}) \subset g(\mathbb{U})$.

Subordination plays an important role in univalent function theory, and this concept was first introduced by Lindelöf, but Littlewood [3, 4] contributed remarkably to this field. Differential subordination is actually the generalized version of differential inequalities with real variables. Many researchers contributed in the work related to differential subordinations. Historical developments in the field of differential subordination are briefly described by Miller and Mocanu in [5].

The advancement in the field of differential subordination starts with the usage of univalent functions. It was noticed in an article by Miller et al. [6] in 1974. Furthermore, many developments in this field have been achieved with the usage of differential subordinations in past fifty years. Differential inequality was a very well-known concept of real variables, and to study it in terms of complex variables, Miller and Mocanu [7] in 1981 were the first ones to introduce the idea of differential subordination. The contribution of Ruscheweyh and Singh [8] and Ruscheweyh and Wilken [9] is also of great importance in this field. Wellknown Jack's lemma [10] has brought the advancements in differential subordinations. Dziok [11] worked on some of the applications of Jack's lemma. The research work carried out by Ma and Minda [12] in the function theory is worth to mention here, as they introduced the analytic function Φ , which satisfies the conditions of normalization $\Phi(0) = 0$ and $\Phi'(0) > 1$ having real part positive. The authors in [12] utilized the function Φ and introduced the subclass $S^*(\Phi)$ of starlike functions as follows:

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$$\mathcal{S}^{*}(\Phi) = \left\{ f \in \mathscr{A} \colon \frac{\varsigma f'(\varsigma)}{f(\varsigma)} \prec \Phi(\varsigma); \quad \varsigma \in \mathbb{U} \right\}.$$
(2)

The idea presented in [12] is very useful, and it helped many researchers for further studies in this direction. Ali et al. [13, 14] worked on differential subordination for sufficiency criteria of Janowski starlikeness and evaluated several differential subordinations such as $1 + \gamma \varsigma$ $(p'(\varsigma)/p^n(\varsigma))$ and found $p(\varsigma) \prec \sqrt{1+\varsigma}$. Also, Ravichandran et al. [15] used this concept to find the sufficient conditions for starlikeness of Bernoulli's lemniscate and Janowski functions. Sharma et al. [16] studied the differential subordinations to prove the starlikeness associated with cardioid domain and Halim et al. [17] introduced the concept for limacon domain.

Jackson [18, 19] was the one who introduced the q-derivatives and q-integrals. After Jackson, Srivastava was amongst the pioneers to contribute in the q-calculus for its usage for analytic functions and their subclasses. Not only this, but he also applied q-hypergeometric function in the functions theory. All these contributions are comprised in his book (pp. 347 in [20]). Ismail et al. [21] contributed in the q-calculus for the study of starlike functions. Anastassiu and Gal [22, 23] also played their part in the development of complex variables with q-generalization. Purohit et al. [24] have used fractional q-calculus operators to apply subordination conditions on the class of non-Bazilevic functions. Sahoo and Agrawal [25] worked on starlike functions in q-calculus and extended the idea of q-starlikeness for particular subclasses of starlike functions. The involvement of q-derivative in the class $\mathcal{S}^*(\Phi)$ gave the formation of following subclass $\mathcal{S}_{a}^{*}(\Phi)$ of starlike functions which was introduced by Aouf and Seoudy [26].

$$\mathcal{S}_{q}^{*}(\Phi) = \left\{ f \in \mathscr{A} \colon \frac{\varsigma D_{q} f(\varsigma)}{f(\varsigma)} \prec \Phi(\varsigma); \quad \varsigma \in \mathbb{U} \right\}.$$
(3)

The class described above has drawn the attention of many researchers. Replacing $\Phi(\varsigma)$ with different functions such as Janowski, lemniscate of Bernoulli, cardioid, and limacon, the researchers got the new directions to the study. Srivastava et al. [27] studied q-derivatives to find the relation between different classes of q-starlike functions related to Janowski function. Srivastava et al. [28] introduced the class of q-starlike functions by using general conic domains. They also obtained the bounds on Hankel and Toeplitz determinants for *q*-starlike functions and continued working par excellence. They produced unmatchable results that worked as great motivation for many researchers worldwide. To have an idea of their remarkable work, one can see [29–31], [20, 27, 28, 32–43]. Contributions of Haq et al. [44] and Zainab et al. [45] are also worth to mention. They studied q-analogue of differential subordinations for star-like functions related to limacon and cardioid domains, and Janowski functions. The q-derivative is the foundation of all this work in q-analogue, and it is defined as follows.

The *q*-derivative of a complex valued function *f*, defined in the domain U, is given as follows:

$$\left(D_q f \right)(\varsigma) = \begin{cases} \frac{f(\varsigma) - f(q\varsigma)}{(1-q)\varsigma}, & \varsigma \neq 0, \\ \\ f'(0), & \varsigma = 0, \end{cases}$$

$$(4)$$

where 0 < q < 1. This implies the following:

$$\lim_{q \to 1^{-}} \left(D_q f \right)(\varsigma) = \lim_{q \to 1^{-}} \frac{f(\varsigma) - f(q\varsigma)}{(1-q)\varsigma} = f'(\varsigma), \tag{5}$$

on the assumption that the function f is differentiable in \mathbb{U} . The q-derivative $D_q f$ of an analytic function f has Taylor series of the form

$$\left(D_q f\right)(\varsigma) = \sum_{n=0}^{\infty} \left[n\right]_q a_n \varsigma^{n-1},\tag{6}$$

where

$$[n]_{q} = \begin{cases} \frac{1-q^{n}}{1-q}, & n \in \mathbb{C}, \\ & & \\ \sum_{k=0}^{n-1} q^{k}, & n \in \mathbb{N}. \end{cases}$$
(7)

For more details about q-derivative and recent work on it, we refer the readers to [29–31], [20, 27, 28, 32–43]. In addition, the q-analogue of Jack's lemma has played a vital role in this paper which states as follows.

Lemma 1 (see [46]). Let ϖ be an analytic function in \mathbb{U} with $\varpi(0) = 0$. For maximum of ϖ on $|\varsigma| = 1$ at $\varsigma_0 = ae^{i\theta}$, where $\theta \in [-\pi, \pi]$ and 0 < q < 1, then we have

$$\varsigma_0 D_q \mathfrak{O}(\varsigma_0) = m \mathfrak{O}(\varsigma_0), \tag{8}$$

where $m \in \mathbb{R}$ with $m \ge 1$.

2. Main Results

Theorem 1. Assume that

$$|\gamma| \ge \frac{(A-B)(\sqrt{2} + \sqrt{3-q})}{(1-|B|)}, \quad -1 < B < A \le 1.$$
(9)

Consider an analytic function h on \mathbb{U} with h(0) = 1 which satisfies

$$1 + \gamma \varsigma D_q h(\varsigma) \prec \frac{1 + A\varsigma}{1 + B\varsigma}, \quad \varsigma \in \mathbb{U}.$$
(10)

Also, suppose

$$1 + \gamma \varsigma D_q h(\varsigma) = \frac{1 + A\overline{\omega}(\varsigma)}{1 + B\overline{\omega}(\varsigma)}, \quad \varsigma \in \mathbb{U}.$$
 (11)

Here, ϖ is an analytic function in \mathbb{U} such that $\varpi(0) = 0$. Then, we have

$$h(\varsigma) \prec \sqrt{1+\varsigma}.\tag{12}$$

Proof. Suppose that

$$p(\varsigma) = 1 + \gamma \varsigma D_q h(\varsigma), \tag{13}$$

where *p* is analytic, and we have p(0) = 1. Also, consider that

$$h(\varsigma) = \sqrt{1 + \mathfrak{O}(\varsigma)}.$$
 (14)

Now, we prove that $|\varpi(\varsigma)| < 1$, where

$$\varpi(\varsigma) = \frac{p(\varsigma) - 1}{A - Bp(\varsigma)}.$$
(15)

Using (13) and (14), we obtain

$$p(\varsigma) = 1 + \gamma\varsigma \frac{D_q \bar{\omega}(\varsigma)}{\sqrt{1 + \bar{\omega}(\varsigma)} + \sqrt{1 + \bar{\omega}(\varsigma) - \varsigma D_q \bar{\omega}(\varsigma)(1 - q)}}$$
(16)

Also, we have

$$\left|\frac{p(\varsigma)-1}{A-Bp(\varsigma)}\right| = \left|\frac{\gamma\varsigma D_q \mathfrak{D}(\varsigma)}{(A-B)\left[\sqrt{1+\mathfrak{D}(\varsigma)} + \sqrt{1+\mathfrak{D}(\varsigma) - \varsigma D_q \mathfrak{D}(\varsigma)(1-q)}\right] - B\gamma\varsigma D_q \mathfrak{D}(\varsigma)}\right|.$$
(17)

Consider a point $\varsigma_0 \in \mathbb{U}$ such that

$$\max_{|\varsigma| \le |\varsigma_0|} |\varpi(\varsigma)| = |\varpi(\varsigma_0)| = 1.$$
(18)

Now, by using Lemma 1, we have $\varsigma_0 D_q \mathfrak{Q}(\varsigma_0) = m\mathfrak{Q}(\varsigma_0), m \ge 1$. Now, consider that $\mathfrak{Q}(\varsigma_0) = e^{i\theta}, \quad \theta \in [-\pi, \pi]$; then, for $\varsigma_0 \in \mathbb{U}$, we obtain

$$\left|\frac{p(\varsigma_{0})-1}{|A-Bp(\varsigma_{0})|}\right| = \left|\frac{\gamma\varsigma_{0}D_{q}\varpi(\varsigma_{0})}{|(A-B)\left[\sqrt{1+\varpi(\varsigma_{0})}+\sqrt{1+\varpi(\varsigma_{0})}-\varsigma_{0}D_{q}\varpi(\varsigma_{0})(1-q)}\right]-B\gamma\varsigma_{0}D_{q}\varpi(\varsigma_{0})|}\right|,$$

$$\geq \frac{|\gamma|m}{(A-B)\left[\sqrt{|1|+|e^{i\theta}|}+\sqrt{|1|+|e^{i\theta}|+|me^{i\theta}(1-q)|}\right]+|B||\gamma|m},$$

$$= \frac{|\gamma|m}{(A-B)\left[\sqrt{2}+\sqrt{2+m(1-q)}\right]+|B||\gamma|m}.$$
(19)

Consider a new function

Then,

$$\Xi(m) = \frac{|\gamma|m}{(A-B)[\sqrt{2} + \sqrt{2 + m(1-q)}] + |B||\gamma|m}.$$
 (20)

$$\Xi'(m) = \frac{|\gamma|[(A-B)\{\sqrt{2} + \sqrt{2+m(1-q)}\}] - |\gamma|m[((A-B)(1-q)/2\sqrt{2+m(1-q)})]}{[(A-B)\{\sqrt{2} + \sqrt{2+m(1-q)}\} + |B||\gamma|m]^2} > 0.$$
(21)

Above expression represents that the function Ξ has increasing behavior, so we have its minimum value at m = 1 and

$$\Xi(1) = \frac{|\gamma|}{(A-B)\left[\sqrt{2} + \sqrt{3-q}\right] + |B||\gamma|}.$$
 (22)

So, we conclude that

$$\left|\frac{p(\varsigma_0)-1}{A-Bp(\varsigma_0)}\right| \ge \frac{|\gamma|}{(A-B)\left[\sqrt{2}+\sqrt{3-q}\right]+|B||\gamma|}.$$
 (23)

Now, from (9), we have

$$\left|\frac{p(\varsigma_0) - 1}{A - Bp(\varsigma_0)}\right| \ge 1,\tag{24}$$

Since this result contradicts (10), therefore, $|\varpi(\varsigma)| < 1$, which completes the proof.

By taking $h(\varsigma) = (\varsigma D_q f(\varsigma)/f(\varsigma))$, we deduce the following result.

Corollary 1. Let $|\gamma| \ge ((A - B)(\sqrt{2} + \sqrt{3 - q})/(1 - |B|)),$ $-1 < B < A \le 1$ and $f \in \mathcal{A}$, satisfy the subordination

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$$1 + \gamma \varsigma D_q \left(\frac{\varsigma D_q f(\varsigma)}{f(\varsigma)}\right) \prec \frac{1 + A\varsigma}{1 + B\varsigma}.$$
 (25)

Then, $f \in \mathcal{S}_q^*(\sqrt{1+\varsigma})$.

Theorem 2. Assume that

$$|\gamma| \ge \frac{\sqrt{2} (A - B) (\sqrt{2} + \sqrt{3 - q})}{(1 - |B|)}, \quad -1 < B < A \le 1.$$
 (26)

Consider an analytic function h on \mathbb{U} with h(0) = 1 which satisfies

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h(\varsigma)} \prec \frac{1 + A\varsigma}{1 + B\varsigma}, \quad \varsigma \in \mathbb{U}.$$
 (27)

Also, suppose

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h(\varsigma)} = \frac{1 + A\overline{\omega}(\varsigma)}{1 + B\overline{\omega}(\varsigma)}, \quad \varsigma \in \mathbb{U},$$
(28)

where ϖ is analytic function on \mathbb{U} with $\varpi(0) = 0$. Then,

$$h(\varsigma) \prec \sqrt{1+\varsigma}.\tag{29}$$

Proof. We define a function

$$p(\varsigma) = 1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h(\varsigma)},\tag{30}$$

where *p* is analytic and p(0) = 1. Now, consider that

$$h(\varsigma) = \sqrt{1 + \omega(\varsigma)}.$$
 (31)

To obtain the result, we have to show that $|\varpi(\varsigma)| < 1$. Using (30) and (31), we obtain the result

$$p(\varsigma) = 1 + \gamma \varsigma \frac{D_q \mathfrak{D}(\varsigma)}{\sqrt{1 + \mathfrak{D}(\varsigma)} \left[\sqrt{1 + \mathfrak{D}(\varsigma)} + \sqrt{1 + \mathfrak{D}(\varsigma) - \varsigma D_q \mathfrak{D}(\varsigma)(1 - q)}\right]}.$$
(32)

Also, we have

$$\left|\frac{p(\varsigma)-1}{A-Bp(\varsigma)}\right| = \left|\frac{\gamma\varsigma D_q \mathfrak{D}(\varsigma)}{(A-B)\sqrt{1+\mathfrak{D}(\varsigma)}\left[\sqrt{1+\mathfrak{D}(\varsigma)} + \sqrt{1+\mathfrak{D}(\varsigma)-\varsigma D_q \mathfrak{D}(\varsigma)(1-q)}\right] - B\gamma\varsigma D_q \mathfrak{D}(\varsigma)}\right|.$$
(33)

Consider a point $\varsigma_0 \in \mathbb{U}$ such that

$$\max_{|\varsigma| \le |\varsigma_0|} |\varpi(\varsigma)| = |\varpi(\varsigma_0)| = 1.$$
(34)

Now, by using Lemma 1, we have $\varsigma_0 D_q \mathfrak{O}(\varsigma_0) = m\mathfrak{O}(\varsigma_0), m \ge 1$. Now, consider that $\mathfrak{O}(\varsigma_0) = e^{i\theta}, \quad \theta \in [-\pi, \pi]$; then, for $\varsigma_0 \in \mathbb{U}$, we obtain

$$\left| \frac{p(\varsigma_{0}) - 1}{|A - Bp(\varsigma_{0})|} \right| = \left| \frac{\gamma m e^{i\theta}}{(A - B)\sqrt{1 + e^{i\theta}} \left[\sqrt{1 + e^{i\theta}} + \sqrt{1 + e^{i\theta} - m e^{i\theta}(1 - q)} \right] - B\gamma m e^{i\theta}} \right|, \qquad (35)$$

$$\geq \frac{|\gamma|m}{(A - B)\sqrt{|1| + |e^{i\theta}|} \left[\sqrt{|1| + |e^{i\theta}|} + \sqrt{|1| + |e^{i\theta}| + |m e^{i\theta}(1 - q)|} \right] + |B||\gamma|m}, \qquad (35)$$

$$= \frac{|\gamma|m}{(A - B)\sqrt{2} \left[\sqrt{2} + \sqrt{2 + m(1 - q)} \right] + |B||\gamma|m}.$$

Consider

$$\Xi_{1}(m) = \frac{|\gamma|m}{(A-B)\sqrt{2}\left[\sqrt{2} + \sqrt{2+m(1-q)}\right] + |B||\gamma|m}.$$
(36)

Then,

$$\Xi_{1}'(m) = \frac{|\gamma| \left[(A-B)\sqrt{2} \left\{ \sqrt{2} + \sqrt{2+m(1-q)} \right\} \right] - |\gamma|m[(\sqrt{2} (A-B)(1-q)/2\sqrt{2+m(1-q)})]}{\left[(A-B)\sqrt{2} \left\{ \sqrt{2} + \sqrt{2+m(1-q)} \right\} + |B||\gamma|m]^{2}} > 0.$$
(37)

Above expression represents that function Ξ_1 has increasing behavior, so we have its minimum value at m = 1 and

$$\Xi_1(1) = \frac{|\gamma|}{(A-B)\sqrt{2}\left[\sqrt{2} + \sqrt{3-q}\right] + |B||\gamma|}.$$
 (38)

So, we conclude that

$$\left|\frac{p(\varsigma_0)-1}{A-Bp(\varsigma_0)}\right| \ge \frac{|\gamma|}{(A-B)\sqrt{2}\left[\sqrt{2}+\sqrt{3-q}\right]+|B||\gamma|}.$$
 (39)

Now, from (26), we have

$$\left|\frac{p\left(\varsigma_{0}\right)-1}{A-Bp\left(\varsigma_{0}\right)}\right|\geq1,\tag{40}$$

which contradicts (27), and hence, $|\varpi(\varsigma)| < 1$, which completes the proof.

By taking $h(\varsigma) = (\varsigma D_q f(\varsigma) / f(\varsigma))$, we deduce the following result.

Corollary 2. Let $|\gamma| \ge \sqrt{2} ((A - B)(\sqrt{2} + \sqrt{3 - q})/(1 - |B|))$, $-1 < B < A \le 1$ and $f \in \mathcal{A}$, which satisfies the subordination

$$1 + \gamma \varsigma \left(\frac{f(\varsigma)}{\varsigma D_q f(\varsigma)}\right) D_q \left(\frac{\varsigma D_q f(\varsigma)}{f(\varsigma)}\right) \prec \frac{1 + A\varsigma}{1 + B\varsigma}.$$
 (41)

Then, $f \in \mathcal{S}_q^*(\sqrt{1+\varsigma})$.

Theorem 3. Assume that

$$|\gamma| \ge \frac{2(A-B)(\sqrt{2}+\sqrt{3-q})}{(1-|B|)}, \quad -1 < B < A \le 1.$$
(42)

Consider an analytic function h on U with h(0) = 1 which satisfies

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^2(\varsigma)} \prec \frac{1 + A\varsigma}{1 + B\varsigma}, \quad \varsigma \in \mathbb{U}.$$
(43)

Also, suppose

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^2(\varsigma)} = \frac{1 + A\overline{\omega}(\varsigma)}{1 + B\overline{\omega}(\varsigma)}, \quad \varsigma \in \mathbb{U},$$
(44)

where ϖ is analytic function on \mathbb{U} with $\varpi(0) = 0$. Then,

$$h(\varsigma) \prec \sqrt{1+\varsigma}.\tag{45}$$

Proof. We define a function

$$p(\varsigma) = 1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^2(\varsigma)},\tag{46}$$

where p is analytic, and we have p(0) = 1. Now, consider that

$$h(\varsigma) = \sqrt{1 + \omega(\varsigma)}.$$
(47)

To obtain the result, we have to show that $|\varpi(\varsigma)| < 1$. Using (46) and (47), we obtain the result

$$p(\varsigma) = 1 + \gamma \varsigma \frac{D_q \mathfrak{D}(\varsigma)}{\sqrt{1 + \mathfrak{D}(\varsigma)} \left[\sqrt{1 + \mathfrak{D}(\varsigma)} + \sqrt{1 + \mathfrak{D}(\varsigma) - \varsigma D_q \mathfrak{D}(\varsigma)(1 - q)}\right]}.$$
(48)

Also, we have

$$\left|\frac{p(\varsigma)-1}{A-Bp(\varsigma)}\right| = \left|\frac{\gamma\varsigma D_q \bar{\omega}(\varsigma)}{(A-B)\sqrt{1+\bar{\omega}(\varsigma)}\left[\sqrt{1+\bar{\omega}(\varsigma)} + \sqrt{1+\bar{\omega}(\varsigma) - \varsigma D_q \bar{\omega}(\varsigma)(1-q)}\right] - B\gamma\varsigma D_q \bar{\omega}(\varsigma)}\right|.$$
(49)

Consider a point $\varsigma_0 \in \mathbb{U}$ such that

$$\max_{|\varsigma| \le |\varsigma_0|} |\mathfrak{O}(\varsigma)| = |\mathfrak{O}(\varsigma_0)| = 1.$$
(50)

Now, by using Lemma 1, we have $\varsigma_0 D_q \mathfrak{Q}(\varsigma_0) = m \mathfrak{Q}(\varsigma_0), m \ge 1$. Now, consider that $\mathfrak{Q}(\varsigma_0) = e^{i\theta}, \quad \theta \in [-\pi, \pi]$; then, for $\varsigma_0 \in \mathbb{U}$, we obtain

$$\left|\frac{p(\varsigma_{0})-1}{|A-Bp(\varsigma_{0})|}\right| = \left|\frac{\gamma m e^{i\theta}}{|(A-B)\sqrt{1+e^{i\theta}}\left[\sqrt{1+e^{i\theta}}+\sqrt{1+e^{i\theta}}-m e^{i\theta}(1-q)\right]-B\gamma m e^{i\theta}}\right|,$$

$$\geq \frac{|\gamma|m}{(A-B)\sqrt{|1|+|e^{i\theta}|}\left[\sqrt{|1|+|e^{i\theta}|}+\sqrt{|1|+|e^{i\theta}|+|m e^{i\theta}(1-q)|}\right]+|B||\gamma|m},$$

$$= \frac{|\gamma|m}{(A-B)2[\sqrt{2}+\sqrt{2}+m(1-q)]+|B||\gamma|m}.$$
(51)

Consider a function

Then,

$$\Xi_{2}(m) = \frac{|\gamma|m}{(A-B)2[\sqrt{2} + \sqrt{2 + m(1-q)}] + |B||\gamma|m}.$$
(52)

$$\Xi_{2}'(m) = \frac{|\gamma|[(A-B)2\{\sqrt{2} + \sqrt{2+m(1-q)}\}] - |\gamma|m[(2(A-B)(1-q)/2\sqrt{2+m(1-q)})]}{[(A-B)2\{\sqrt{2} + \sqrt{2+m(1-q)}\} + |B||\gamma|m]^{2}} > 0.$$
(53)

Here, Ξ_2 is clearly an increasing function, so we have its minimum value at m = 1 and

$$\Xi_{2}(1) = \frac{|\gamma|}{2(A-B)[\sqrt{2} + \sqrt{3-q}] + |B||\gamma|}.$$
 (54)

So, we conclude that

$$\left|\frac{p(\varsigma_0) - 1}{A - Bp(\varsigma_0)}\right| \ge \frac{|\gamma|}{2(A - B)[\sqrt{2} + \sqrt{3 - q}] + |B||\gamma|}.$$
 (55)

Now, from (42), we have

$$\left|\frac{p(\varsigma_0) - 1}{A - Bp(\varsigma_0)}\right| \ge 1,\tag{56}$$

which contradict (43), and hence, $|\varpi(\varsigma)| < 1$, which completes the proof.

By taking $h(\varsigma) = (\varsigma D_q f(\varsigma) / f(\varsigma))$, we deduce the following result.

Corollary 3. Let $|\gamma| \ge 2((A - B)(\sqrt{2} + \sqrt{3 - q})/(1 - |B|))$, $-1 < B < A \le 1$ and $f \in \mathcal{A}$, satisfies the subordination

$$1 + \gamma \varsigma \left(\frac{f(\varsigma)}{\varsigma D_q f(\varsigma)}\right)^2 D_q \left(\frac{\varsigma D_q f(\varsigma)}{f(\varsigma)}\right) \prec \frac{1 + A\varsigma}{1 + B\varsigma}.$$
 (57)

Then,
$$f \in \mathcal{S}_q^*(\sqrt{1+\varsigma})$$
.

Theorem 4. Assume that

$$|\gamma| \ge \frac{2\sqrt{2}(A-B)(\sqrt{2}+\sqrt{3-q})}{(1-|B|)}, \quad -1 < B < A \le 1.$$
(58)

Consider an analytic function h on \mathbb{U} with h(0) = 1 which satisfies

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^3(\varsigma)} \prec \frac{1 + A\varsigma}{1 + B\varsigma}, \quad \varsigma \in \mathbb{U}.$$
 (59)

Also, suppose

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^3(\varsigma)} = \frac{1 + A\overline{\omega}(\varsigma)}{1 + B\overline{\omega}(\varsigma)}, \quad \varsigma \in \mathbb{U},$$
(60)

where ϖ is analytic function on \mathbb{U} with $\varpi(0) = 0$. Then,

$$h(\varsigma) \prec \sqrt{1+\varsigma}.\tag{61}$$

Proof. We define a function

$$p(\varsigma) = 1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^3(\varsigma)},\tag{62}$$

where p is analytic, and we have p(0) = 1. Now, consider that

$$h(\varsigma) = \sqrt{1 + \varpi(\varsigma)}.$$
 (63)

To obtain the result, we have to show that $|\overline{\omega}(\varsigma)| < 1$. Using (62) and (63), we obtain

$$p(\varsigma) = 1 + \gamma \varsigma \frac{D_q \mathfrak{D}(\varsigma)}{(1 + \mathfrak{D}(\varsigma))^{(3/2)} \left[\sqrt{1 + \mathfrak{D}(\varsigma)} + \sqrt{1 + \mathfrak{D}(\varsigma) - \varsigma D_q \mathfrak{D}(\varsigma)(1 - q)}\right]}.$$
(64)

Also, we have

$$\left|\frac{p(\varsigma)-1}{|A-Bp(\varsigma)|}\right| = \left|\frac{\gamma\varsigma D_q \mathfrak{D}(\varsigma)}{\left(A-B\right)\left(1+\mathfrak{D}(\varsigma)\right)^{(3/2)} \left[\sqrt{1+\mathfrak{D}(\varsigma)} + \sqrt{1+\mathfrak{D}(\varsigma)-\varsigma D_q \mathfrak{D}(\varsigma)\left(1-q\right)}\right] - B\gamma\varsigma D_q \mathfrak{D}(\varsigma)}\right|.$$
(65)

Consider a point $\varsigma_0 \in \mathbb{U}$ such that

$$\max_{|\varsigma| \le |\varsigma_0|} |\varpi(\varsigma)| = |\varpi(\varsigma_0)| = 1.$$
(66)

Now, by using Lemma 1, we have $\varsigma_0 D_q \mathfrak{Q}(\varsigma_0) = m\mathfrak{Q}(\varsigma_0), m \ge 1$. Now, consider that $\mathfrak{Q}(\varsigma_0) = e^{i\theta}, \quad \theta \in [-\pi, \pi]$; then, for $\varsigma_0 \in \mathbb{U}$, we obtain

$$\left|\frac{p(\varsigma_{0})-1}{|A-Bp(\varsigma_{0})}\right| = \left|\frac{\gamma m e^{i\theta}}{(A-B)(1+e^{i\theta})^{(3/2)} \left[\sqrt{1+e^{i\theta}}+\sqrt{1+e^{i\theta}}-m e^{i\theta}(1-q)\right] - B\gamma m e^{i\theta}}\right|,$$

$$\geq \frac{|\gamma|m}{(A-B)(|1|+|e^{i\theta}|)^{(3/2)} \left[\sqrt{|1|+|e^{i\theta}|}+\sqrt{|1|+|e^{i\theta}|+|m e^{i\theta}(1-q)|}\right] + |B||\gamma|m},$$

$$= \frac{|\gamma|m}{(A-B)2^{(3/2)} \left[\sqrt{2}+\sqrt{2+m(1-q)}\right] + |B||\gamma|m}.$$
(67)

Consider a function

Then,

 $\Xi_{3}(m) = \frac{|\gamma|m}{2^{(3/2)} (A-B) [\sqrt{2} + \sqrt{2 + m(1-q)}] + |B||\gamma|m}.$ (68)

$$\Xi_{3}'(m) = \frac{|\gamma| \left[(A-B)2^{(3/2)} \left\{ \sqrt{2} + \sqrt{2 + m(1-q)} \right\} \right] - |\gamma| m \left[\left(2^{(3/2)} (A-B)(1-q)/2\sqrt{2 + m(1-q)} \right) \right]}{\left[(A-B)2^{(3/2)} \left\{ \sqrt{2} + \sqrt{2 + m(1-q)} \right\} + |B||\gamma|m \right]^{2}} > 0.$$
(69)

Above expression represents that function Ξ_3 has increasing behavior, so we have its minimum value at m = 1 and

$$\Xi_{3}(1) = \frac{|\gamma|}{2^{(3/2)} (A - B) \left[\sqrt{2} + \sqrt{3 - q}\right] + |B||\gamma|}.$$
 (70)

So, we conclude that

$$\left|\frac{p(\varsigma_0) - 1}{A - Bp(\varsigma_0)}\right| \ge \frac{|\gamma|}{2^{(3/2)} (A - B) [\sqrt{2} + \sqrt{3 - q}] + |B||\gamma|}.$$
(71)

Now, from (58), we have

$$\left|\frac{p\left(\varsigma_{0}\right)-1}{A-Bp\left(\varsigma_{0}\right)}\right| \ge 1,\tag{72}$$

which contradicts (59), and hence, $|\omega(\varsigma)| < 1$, which completes the proof.

By taking $h(\varsigma) = (\varsigma D_q f(\varsigma)/f(\varsigma))$, we deduce the following result.

Corollary 5. Let $|\gamma| \ge 2\sqrt{2}((A-B)(\sqrt{2} + \sqrt{3-q})/(1-|B|)), -1 < B < A \le 1$ and $f \in \mathcal{A}$, satisfy the subordination

$$1 + \gamma \varsigma \left(\frac{f(\varsigma)}{\varsigma D_q f(\varsigma)}\right)^3 D_q \left(\frac{\varsigma D_q f(\varsigma)}{f(\varsigma)}\right) \prec \frac{1 + A\varsigma}{1 + B\varsigma}.$$
 (73)

Then, $f \in \mathcal{S}_q^*(\sqrt{1+\varsigma})$.

Theorem 6. Assume that

$$|\gamma| \ge \frac{2^{(n/2)} (A - B) (\sqrt{2} + \sqrt{3 - q})}{(1 - |B|)}, \quad -1 < B < A \le 1.$$
(74)

Consider an analytic function h on \mathbb{U} with h(0) = 1 which satisfies

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^n(\varsigma)} < \frac{1 + A\varsigma}{1 + B\varsigma}, \quad \varsigma \in \mathbb{U}.$$
 (75)

Also, suppose

$$1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^n(\varsigma)} = \frac{1 + A\overline{\omega}(\varsigma)}{1 + B\overline{\omega}(\varsigma)}, \quad \varsigma \in \mathbb{U},$$
(76)

where ϖ is analytic function on \mathbb{U} with $\varpi(0) = 0$. Then,

$$h(\varsigma) \prec \sqrt{1+\varsigma}.\tag{77}$$

Proof. We define a function

$$p(\varsigma) = 1 + \frac{\gamma \varsigma D_q h(\varsigma)}{h^n(\varsigma)},\tag{78}$$

where p is analytic, and we have p(0) = 1. Now, consider that

$$h(\varsigma) = \sqrt{1 + \varpi(\varsigma)}.$$
(79)

To obtain the result, we have to show that $|\varpi(\varsigma)| < 1$. Using (78) and (79), we obtain

$$p(\varsigma) = 1 + \gamma \varsigma \frac{D_q \mathfrak{D}(\varsigma)}{(1 + \mathfrak{D}(\varsigma))^{(n/2)} \left[\sqrt{1 + \mathfrak{D}(\varsigma)} + \sqrt{1 + \mathfrak{D}(\varsigma) - \varsigma D_q \mathfrak{D}(\varsigma)(1 - q)}\right]}.$$
(80)

Also, we have

$$\left|\frac{p(\varsigma)-1}{A-Bp(\varsigma)}\right| = \left|\frac{\gamma\varsigma\left(D_q\mathfrak{D}(\varsigma)/(1+\mathfrak{D}(\varsigma))^{(n/2)}\left[\sqrt{1+\mathfrak{D}(\varsigma)}+\sqrt{1+\mathfrak{D}(\varsigma)}-\varsigma D_q\mathfrak{D}(\varsigma)(1-q)\right]\right)\right|}{A-B\left[1+\gamma\varsigma\left(D_q\mathfrak{D}(\varsigma)/(1+\mathfrak{D}(\varsigma))^{(n/2)}\left[\sqrt{1+\mathfrak{D}(\varsigma)}+\sqrt{1+\mathfrak{D}(\varsigma)}-\varsigma D_q\mathfrak{D}(\varsigma)(1-q)\right]\right)\right]\right|} = \left|\frac{\gamma\varsigma D_q\mathfrak{D}(\varsigma)}{(A-B)(1+\mathfrak{D}(\varsigma))^{(n/2)}\left[\sqrt{1+\mathfrak{D}(\varsigma)}+\sqrt{1+\mathfrak{D}(\varsigma)}-\varsigma D_q\mathfrak{D}(\varsigma)(1-q)\right]-B\gamma\varsigma D_q\mathfrak{D}(\varsigma)}\right|.$$
(81)

Consider a point $\varsigma_0 \in \mathbb{U}$ such that $\max |\varpi(\varsigma)| = |\varpi(\varsigma_0)| = 1.$

$$\max_{|\varsigma| \le |\varsigma_0|} |\varpi(\varsigma)| = |\varpi(\varsigma_0)| = 1.$$
(82)
$$\varsigma_0 D \\ \varpi(\varsigma_0) = 0.$$

Now, by using Lemma 1, we have $\varsigma_0 D_q \mathfrak{O}(\varsigma_0) = m\mathfrak{O}(\varsigma_0), m \ge 1$. Now, consider that $\mathfrak{O}(\varsigma_0) = e^{i\theta}, \quad \theta \in [-\pi, \pi]$; then, for $\varsigma_0 \in \mathbb{U}$, we obtain

$$\begin{aligned} \left| \frac{p(\varsigma_{0}) - 1}{|A - Bp(\varsigma_{0})|} \right| &= \frac{\gamma \varsigma_{0} D_{q} \tilde{\omega}(\varsigma_{0})}{(A - B)(1 + \tilde{\omega}(\varsigma_{0}))^{(n/2)} \left[\sqrt{1 + \tilde{\omega}(\varsigma_{0})} + \sqrt{1 + \tilde{\omega}(\varsigma_{0})} - \varsigma_{0} D_{q} \tilde{\omega}(\varsigma_{0})(1 - q)} \right] - B\gamma \varsigma_{0} D_{q} \tilde{\omega}(\varsigma_{0})}, \\ &= \left| \frac{\gamma m e^{i\theta}}{(A - B)(1 + e^{i\theta})^{(n/2)} \left[\sqrt{1 + e^{i\theta}} + \sqrt{1 + e^{i\theta}} - m e^{i\theta}(1 - q)} \right] - B\gamma m e^{i\theta}} \right|, \\ &\geq \frac{|\gamma|m}{(A - B)(1 + e^{i\theta})^{(n/2)} \left[\left| \sqrt{1 + e^{i\theta}} \right| + \left| \sqrt{1 + e^{i\theta}} - m e^{i\theta}(1 - q)} \right| \right] - |B||\gamma|m}, \\ &= \frac{|\gamma|m}{(A - B)(|1| + |e^{i\theta}|)^{(n/2)} \left[\sqrt{|1| + e^{i\theta}|} + \sqrt{|1 + e^{i\theta}} - m e^{i\theta}(1 - q)|} \right] + |B||\gamma|m}, \\ &\geq \frac{|\gamma|m}{(A - B)(|1| + |e^{i\theta}|)^{(n/2)} \left[\sqrt{|1| + |e^{i\theta}|} + \sqrt{|1| + |e^{i\theta}|} + |m e^{i\theta}(1 - q)|} \right] + |B||\gamma|m}, \\ &= \frac{|\gamma|m}{(A - B)(|1| + |e^{i\theta}|)^{(n/2)} \left[\sqrt{|1| + |e^{i\theta}|} + \sqrt{|1| + |e^{i\theta}|} + |m e^{i\theta}(1 - q)|} \right] + |B||\gamma|m}, \end{aligned}$$

Consider

Then,

$$\Xi_4(m) = \frac{|\gamma|m}{2^{(n/2)} (A-B) [\sqrt{2} + \sqrt{2 + m(1-q)}] + |B||\gamma|m}.$$
(84)

$$\Xi_{4}'(m) = \frac{|\gamma| \left[(A-B)2^{(n/2)} \left\{ \sqrt{2} + \sqrt{2+m(1-q)} \right\} \right] - |\gamma|m \left[\left(2^{(n/2)} (A-B)(1-q)/2\sqrt{2+m(1-q)} \right) \right]}{\left[(A-B)2^{(n/2)} \left\{ \sqrt{2} + \sqrt{2+m(1-q)} \right\} + |B||\gamma|m \right]^{2}} > 0.$$
(85)

Above expression represents that function Ξ_4 has increasing behavior, so we have its minimum value at m = 1 and

$$\Xi_4(1) = \frac{|\gamma|}{2^{(n/2)} (A - B) \left[\sqrt{2} + \sqrt{3 - q}\right] + |B||\gamma|}.$$
 (86)

So, we conclude that

$$\left|\frac{p(\varsigma_{0})-1}{A-Bp(\varsigma_{0})}\right| \geq \frac{|\gamma|}{2^{(n/2)}(A-B)[\sqrt{2}+\sqrt{3-q}]+|B||\gamma|}.$$
(87)

Now, from (74), we have

$$\left|\frac{p(\varsigma_0)-1}{A-Bp(\varsigma_0)}\right| \ge 1,$$
(88)

which contradict (75), and hence, $|\varpi(\varsigma)| < 1$, which completes the proof.

By taking $h(\varsigma) = (\varsigma D_q f(\varsigma)/f(\varsigma))$, we deduce the following result.

Corollary 7. Let $|\gamma| \ge (2^{(n/2)}(A-B)(\sqrt{2} + \sqrt{3-q})/(1-|B|))$, $-1 < B < A \le 1$ and $f \in \mathcal{A}$, satisfy the subordination

$$1 + \gamma \varsigma \left(\frac{f(\varsigma)}{\varsigma D_q f(\varsigma)}\right)^n D_q \left(\frac{\varsigma D_q f(\varsigma)}{f(\varsigma)}\right) \prec \frac{1 + A\varsigma}{1 + B\varsigma}.$$
 (89)

Then, $f \in \mathcal{S}_q^*(\sqrt{1+\varsigma})$.

3. Conclusion

In this article, we have investigated the *q*-differential subordination by using *q*-version of well-known Jack's Lemma. We have found the condition on γ such that $1 + (\gamma \varsigma D_q h(\varsigma)/h^n(\varsigma)) \prec (1 + A\varsigma/1 + B\varsigma)$ implies that $h(\varsigma) \prec \sqrt{1 + \varsigma}$. These results have been utilized to find sufficient conditions for star-like functions related to lemniscate of Bernoulli. This method can further be applied to find sufficient conditions for star-like functions of Ma–Minda type.

Data Availability

No data were used in this article.

Conflicts of Interest

The authors declare that they have no conflicts of interest.

Authors' Contributions

All authors contributed equally and approved the final manuscript.

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