Research Article

Instability Model of a Coal Wall with Large Mining Height under Excavation Unloading Conditions

Lei Li and Feng Zhang

1Business Division of Mining Design of Tiandi Technology Co., Ltd., Beijing 100013, China
2Mining Research Division of China Coal Research Institute, Beijing 100013, China
3State Key Laboratory of Mining Response and Disaster Prevention and Control in Deep Coal Mines, Anhui University of Science and Technology, Huainan, Anhui Province 232001, China
4State Key Laboratory of Water Resource Protection and Utilization in Coal Mining, Beijing 100120, China

Correspondence should be addressed to Lei Li; lilei@tdkcsj.com

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On the basis of the stress field characteristics of surrounding rocks at a coal wall on a working face with a large mining height, the theories of unloading rock mass mechanics and fracture mechanics were used to establish a model of the excavation unloading field effect of the coal wall, and its instability mechanism under the action of unloading stress field was analyzed. Results show that the coal mining process is the unloading process of coal and rock masses, and the stress field of surrounding rocks at the coal wall turns into an unloading stress field that consists of original and unloading stresses. Under the action of unloading stress field, cracks in the coal wall will undergo instability, propagation, and combination in the form of composite-type cracks and will gradually evolve into a wedge structure. The wedge stability is inversely related to roof pressure $P_0$, unloading force $T$, and intersection angle $\phi$ of structural planes. Elevating the wall-supporting force $P$, the initial supporting force of supports on the working face and the cohesion $C$ of coal body can effectively control the occurrence of coal wall caving accidents and contribute to the safe mining of working faces with a large mining height.

1. Introduction

Fully mechanized underground coal mining with a large mining height is the development direction of thick coal mining in China. However, enlarging the mining height aggravates the probability of coal wall caving on the working face [1–3] and restricts the application of fully mechanized coal mining technology in thick coal seam. Current methods of theoretically analyzing coal wall caving causes and mechanism, such as crack-layer plate structural model [4–6], compressive bar structural model [7, 8], wedge stability model [9], mechanical model of coal wall sliding surface [10], or coal wall shear failure model [11], neglect the influence of the coal mining-induced unloading effect. The theory of unloading rock mass mechanics states that the excavation process of rock and earth masses is actually a mechanical unloading process of rock masses [12]. Jointed rock masses in the loading process generally have good mechanical properties, but rock masses are turned from compressive state into tensile state under excavation unloading, and their mechanical properties will be rapidly degraded; as a result, the mechanical characteristics of rock masses under unloading condition are inconsistent with those under loading condition [13–15]. Therefore, the coal rock mass near the coal wall formed by coal seam excavation is analyzed as an unloading rock mass, which conforms to the characteristics of the engineering rock mass, and it has practical engineering significance. On the basis of the theories of unloading rock mass and fracture mechanics, the fracture instability mechanism of a coal wall under the action of unloading stress field was investigated, the fracture instability criterion for the coal wall and the influencing factors
of its instability were acquired, and a new framework of using the theory of unloading rock mass mechanics to analyze coal wall fracture instability was opened.

2. Characteristics of Coal Excavation-Induced Unloading Field Effect

2.1. Principle of Unloading Analysis. The theory of unloading rock mass mechanics posits that excavation unloading effect is equivalent to applying a reverse tensile stress in the rock mass under initial stress. Thus, unloading stress can be equivalent to a tensile stress that acts upon the initial rock mass, and its maximum value is \( \sigma_0 + R_t \) (\( \sigma_0 \) is the initial stress, \( R_t \) is the tensile strength of rock mass, and \( \sigma_0 + R_t \) is the equivalent tensile strength of rock mass). The unloading problem of rock mass during mining can be decomposed into the joint action model of initial and mining unloading stresses, as shown in Figure 1, where \( \Delta \sigma \) is the unloading stress.

2.2. Stress Field of the Coal Body nearby the Coal Wall and Its Failure. In coal excavation process, the coal body nearby the coal wall will also bear the action of the excavation-induced unloading stress in addition to the stress of the primary rock. The stress condition is shown in Figure 2.

Given that original cracks or defects exist in the coal body, the quality of the coal nearby the coal wall will be gradually degraded under the unloading stress field that consists of primary rock stress (compressive stress) and unloading stress (tensile stress). After the excavation of the roadway, the original rock stress is redistributed, and the stress concentration appears in the surrounding rock of the roadway, which generally does not exceed 5 times the radius of the roadway. For the coal body nearby the coal wall of the working face with a large mining height, the coal wall will easily experience tensile or shear failure under the joint action of compressive and tensile stresses. When lateral confining pressure is moderate and lower than brittle-plastic transition value, the coal wall can easily undergo dislocation or shear failure [5, 16]. The occurrence of the above-mentioned failure mode is closely related to crack propagation and coalescence.

\[ K_I = \left( \Delta \sigma \sin^2 \beta - \sigma \right) \sqrt{\pi a}, \]
\[ K_{II} = \frac{\Delta \sigma \sqrt{\pi a} \sin \beta \cos \beta}{2}. \]

According to the theory of the maximum circumferential stress in fracture mechanics [18–20], the breaking angle \( \theta_0 \) of the composite-type crack satisfies the following condition:
\[ k_1 \sin \theta_0 + k_{II} (3 \cos \theta_0 - 1) = 0. \]

Its fracture criterion is
\[ K_I \cos^2 \theta_0 - \frac{3}{2} K_{II} \sin \theta_0 \cos \theta_0 = K_{IC}. \]

We set \( \alpha = K_I/K_{II} \). Then, we have
\[ \alpha = \frac{\Delta \sigma \sin^2 \beta - \sigma}{\Delta \sigma \sin \beta \cos \beta}. \]

From equation (2), we can obtain the following:
\[ \cos \theta_0 = \frac{3 \pm \sqrt{\alpha^4 + 8 \alpha^2}}{\alpha^2 + 9}. \]

From comprehensive formulas (3), (4), and (5), we can get
\[ \frac{\alpha^3 + 12 \alpha \pm \sqrt{\alpha^4 + 8 \alpha^2}}{2(\alpha^2 + 9)} \]
\[ \frac{3}{2} \left( 1 - \frac{9 \pm 3 \sqrt{\alpha^4 + 8 \alpha^2}}{\alpha^2 + 9} \right) \]
\[ \left( \pm \sqrt{\alpha^2 + 12 \pm \sqrt{\alpha^4 + 8 \alpha^2}} \right) \]
\[ \frac{K_{IC}}{\Delta \sigma \sqrt{\pi a} \sin \beta \cos \beta}. \]
We set $m = K_{IC}/\Delta\alpha \sqrt{\pi a} \sin \beta \cos \beta$ and $0 < \theta_0 < \pi/2$. Then, equation (8) of $\alpha$ about $m$ can be solved, and equation (4) can be combined to determine the relationships among ultimate compressive stress $\sigma$, unloading stress $\Delta\sigma$ (tensile stress), crack angle $\beta$, fracture toughness $K_{IC}$, and initial crack length $a$, as shown in equation (9).

$$
\alpha = \frac{m}{3} - \frac{2^{1/3} \left(30m^2 - m^4\right)}{3m \left(-216m^2 + 153m^4 + 2m^6 + 3\sqrt{3}\sqrt{1728m^4 + 1552m^6 + 435m^8 + 36m^{10}}\right)^{1/3}}$
$$

$$
\sigma = \Delta\sigma \left\{ \frac{\sin \beta - m}{3} - \frac{2^{1/3} \left(30m^2 - m^4\right)}{3m \left(-216m^2 + 153m^4 + 2m^6 + 3\sqrt{3}\sqrt{1728m^4 + 1552m^6 + 435m^8 + 36m^{10}}\right)^{1/3}} \right\} \sin \beta,
$$

When the crack angle $\beta$ and the material fracture toughness $K_{IC}$ are determined, the ultimate compressive stress $\sigma$ declines with the increase in the unloading stress $\Delta\sigma$. With the continuous advancement of the working face during the coal mining process, the stress in the coal body in front of the working face is gradually concentrated. The unloading stress is also gradually enlarged. When the compressive and unloading stresses satisfy equation (9), the crack in the jointed coal body will undergo instability, propagation, and combination. This condition will lead to a coal wall caving event during operation under a large mining height. When the composite-type crack in the coal body is propagated, its propagation direction will not be along the direction of the dip angle of original crack any longer, but it will present propagation at the angle of $\theta_0$. Therefore, the instability form of the coal body at the coal wall with a large quantity of cracks is not a simple layer-crack plate structure but an irregular polyhedron of a certain dip angle. This deduction can be verified through the characteristic analysis and statistics of many field caving accidents.

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**Figure 2:** Unloading stress field in the coal body at the coal wall. Force diagram of the (a) coal body and (b) coal units at the coal wall on the working face.

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3. Instability Criterion for Wedge Structure at the Coal Wall

The characteristic statistics of a large number of field coal wall caving accidents indicates that the geometric characteristics of the irregular polyhedron formed by instability, propagation, and combination of composite-type cracks can be simplified into a wedge structural model [9, 21]. The stability of the coal wall can be analyzed through the stability criterion for the coal wall wedge. In the figure, planes I and II are the structural planes that form after crack propagation and combination in the coal body. The line of intersection between these planes is BD, and a wedge with quadrangular points ABCD is formed.

The dead weight of the wedge is set as \( G \), the roof pressure as \( P_0 \), the unloading force as \( T \), the horizontal wall-supporting force provided by wall-supporting plates as \( P_h \), and the structural plane intersection angle as \( \phi \). The line of intersection BD is taken as x-axis, and the vertical line of the outward intersection is taken as y-axis to establish the coordinate system shown in Figure 5(b).

The force \( N \) perpendicular to the line of intersection BD between structural planes is

\[
N = (P_0 + G) \cos \phi + (P_h - T) \sin \phi. \tag{10}
\]

The force \( \tau \) parallel to the line of intersection BD between structural planes is

\[
\tau = (P_0 + G) \sin \phi - (P_h - T) \cos \phi. \tag{11}
\]

\( N \) is decomposed into \( N_1 \) perpendicular to structural plane I, and \( N_{II} \) is perpendicular to structural plane II. We assume that the included angles between vertical plane of line of intersection BD, which passes through structural planes I and II, and two structural planes are \( \theta_1 \) and \( \theta_2 \). The sliding direction of wedge is also assumed parallel to the line of intersection of the two structural planes, as shown in Figures 5(c) and 5(d). Then, it has

\[
N_1 = N \sin \theta_1, \quad N_{II} = N \sin \theta_2. \tag{12}
\]

The wedge stability coefficient \( K \) can be expressed as

\[
K = \frac{f_1 N_1 + f_{II} N_{II} + C_1 S_1 + C_{II} S_{II}}{(P_0 + G) \sin \phi - (P_h - T) \cos \phi}, \tag{13}
\]

where \( f_1 \) and \( f_{II} \) are the sliding coefficients of structural planes I and II, respectively; \( C_1 \) and \( C_{II} \) are their cohesion coefficients; and \( S_1 \) and \( S_{II} \) are the areas of their sliding surfaces.

By combining equations (10), (12), and (13), the expression of stability coefficient \( K \) is determined as in equation (14). When \( K < 1 \), the wedge at the coal wall will undergo instability failure to cause coal wall caving. When \( K \geq 1 \), the wedge will not experience any instability failure or the caving phenomenon.

\[
K = \frac{f_1 [(P_0 + G) \cos \phi + (P_h - T) \sin \phi] \sin \theta_1 + f_{II} [(P_0 + G) \cos \phi + (P_h - T) \sin \phi] \sin \theta_2 + C_1 S_1 + C_{II} S_{II}}{(P_0 + G) \sin \phi - (P_h - T) \cos \phi}. \tag{14}
\]

In investigating the change relations of wedge stability with roof pressure, unloading force, and intersection angle between structural planes, we assume \( \theta_1 = \theta_2 \), \( f_1 = f_{II} \), \( C_1 = C_{II} \), and \( S_1 = S_{II} \). Then, equation (14) can be simplified into the following form:

\[
K = \frac{2 f_1 [(P_0 + G) \cos \phi] \sin \theta_1 + 2 C_1 S_1}{(P_0 + G) \sin \phi - (P_h - T) \cos \phi}. \tag{15}
\]

A single-factor analysis by assigning values to the influencing parameters of wedge stability shows that, as roof pressure \( P_0 \), unloading force \( T \), and intersection angle \( \phi \) between structural planes increase, the wedge stability is degraded. Moreover, the influence degrees of unloading force \( T \) and intersection angle \( \phi \) are high, as shown in Figure 6.

As the wall-supporting force \( P_h \) and cohesion \( C \) of structural planes increase, wedge stability is also enhanced [9]. However, as the coal seam is continuously mined forward, the coal quality at the coal wall will be continuously degraded due to the existence of the excavation unloading effect, and its cohesion and internal frictional angle are continuously decreased, which further aggravates the wedge instability. Thus, on a working face with a large mining height, elevating wall-supporting force \( P_h \), cohesion \( C \) of coal body, and initial supporting force of supports on the working face can improve the wedge stability.
4. Stability Control Technology of the Coal Wall with a Large Mining Height

The wedge instability mechanism at the coal wall indicates that improving the wedge stability helps enhance the stability of the coal wall and control coal wall caving. Therefore, the following measures can be taken to control coal wall caving in engineering practice:

Wall-supporting force \( P_h \) should be elevated. The stability of the coal wall on the working face with a large mining height, especially for "three-soft" coal seam, is poor due to the unloading effect. However, enlarging the wall-supporting force can generate a certain compensating and weakening effect on the unloading force. Thus, after the coal is cut through a coal machine, the wall-supporting plate should be timely opened to guarantee wedge stability.

Cohesion \( C \) of the coal body at the coal wall should be elevated. Coal seam grouting or water injection can improve coal properties, increase coal wall cohesion, enlarge internal frictional angle, and enhance antishear ability and overall stability.

Initial supporting force of supports should be elevated. Initial supporting force is the force used by supports to proactively support the roof. The supports should be timely lifted after coal cutting and support advancing, and roof-contacted filling should be carried out to ensure sufficient initial supporting force, effectively mitigate rotation and subsidence of roof strata, keep peak supporting pressure away from the coal wall, and relieve coal wall pressure \( P_0 \). Accordingly, the wedge stability of coal wall can be controlled.

The advancement of the working face should be properly accelerated. The accelerated advancement of the working face shortens the exposure time of the coal wall and relieves the influence degree of the coal excavation unloading effect. Meanwhile, it also reduces the action time of the supporting pressure on the coal wall and the degree of coal damage and degradation. Accordingly, the coal wall caving degree can be decreased.

5. Conclusions

The coal mining process is the unloading process of coal and rock masses. The characteristics of coal excavation-induced unloading stress field were analyzed using the theory of unloading rock mass mechanics. The results showed that the stress field of surrounding rocks at the coal wall would be turned into an unloading stress field that consisted of original and unloading stresses.
Under the action of unloading stress field, the cracks in the coal wall would undergo instability, propagation, and combination in the form of composite-type cracks and be gradually evolved into a wedge structure. Meanwhile, the propagation criterion for composite-type cracks under the existence of unloading stress was given in accordance with the theory of fracture mechanics.

The wedge instability criterion of the coal wall was obtained by establishing a mechanical model of the wedge stability of the coal wall. Specifically, when $K < 1$, the coal wall wedge would experience instability failure to cause coal wall caving. In addition, the wedge stability would be degraded, and the coal wall could easily undergo caving with the increase in roof pressure $P_0$, unloading force $T$, and intersection angle $\varphi$ between structural planes.

The stability of the coal wall could be improved by elevating the wall-supporting force $P_h$, the coal cohesion $C$, and the initial supporting force and by properly accelerating the advancement of the working face to provide a guarantee for safe mining of working faces with a large mining height.

Data Availability

The data underlying the findings of this paper are publicly available.

Conflicts of Interest

The authors declare no conflicts of interest.

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