Numerical Investigation on the Ground Response of a Gob-Side Entry in an Extra-Thick Coal Seam

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This study was aimed at the large deformation phenomenon of rock mass surrounding the gob-side entry driven in a 20 m extra-thick coal seam. Taking tailgate 8211 as the engineering background, a numerical investigation was employed to analyze the deformation law of the gob-side entry. The study results are as follows. (1) Because the immediate roof was composed of weak coal mass with a thickness of 17 m, the roof coal mass was vulnerable to fail with the effect of overlying strata pressure; thus, a visual subsidence of roof coal mass with a maximum convergence of 800 mm was observed in the field. (2) The bearing capacity of the coal pillar was significantly less than that of the panel rib, resulting in the pillar failing more easily under the ground pressure and then generating large-scale squeezing deformation. (3) The roof and panel rib were in a state of shear failure with a failure depth of about 5 m. The coal pillar was entirely in a state of plastic failure. (4) A support scheme including an asymmetric anchor beam truss, roof angle anchor cable, and anchor cable combination structure was proposed. The field work confirmed that this support scheme could efficiently control the deformation and failure of the rock mass surrounding the gob-side entry. This study provides the theoretical basis and technical support for the control of rocks surrounding the gob-side entry in similar conditions.

1. Introduction

Thick and extra-thick coal seam resources are rich in China, and their reserves and production account for about 45% of the total [1]. Currently, thick and extra-thick coal seams have become the main coal seam for coal mining in China. Gob-side entry is the most commonly used mining mode in thick and extra-thick coal seams, which mostly retains about 20 to 50 m wide coal pillars between adjacent panels to perform coal mining. In recent years, gob-side entry retained with 6–10 m wide coal pillar has begun to be popularized and applied in the thick coal seams [2, 3]. However, due to the high abutment stress induced by mining activities, the roof and two sides of gob-side entries are prone to large-scale deformation and failure, as well as to vicious accidents, such as roof fall and rib spalling [4–7]. Thus, the stability control of gob-side entry has become a key factor restricting the high yield, efficiency, and safe mining of extra-thick coal seams.

In recent years, scholars have conducted a lot of beneficial research on the ground stability of gob-side entry in extra-thick coal seams. Zhang et al. presented a comprehensive field investigation of the ground response of a gate road subjected to high stress induced by extracting a 17 m thick coal seam [8]. Yu et al. pointed out failure modes of the gob-side entry and studied the influence of the failure structure on the stability of coal pillars [9]. Li et al. conducted a research into the balance conditions between the key rock blocks above the gob-side entry and proposed the entry support resistance quantitatively [10]. Shen et al. pointed out that the middle part of the roof is the main part to control the surrounding rock, and three kinds of targeted control technologies were put forward [11]. Feng et al. mentioned that strengthening coal pillars with high-strength bolts is of great significance to surrounding rock stability [12]. However, their studies suffer various limitations. The previous studies are mainly based on specific geological conditions. In reality, due to the complexity and difference of the geological conditions of various thick coal seams bases, the corresponding deformation and failure laws as well as its control method of the gob-side entry show great
The thickness of the #3 coal seam was 5.19 m, and that of #5 coal seam was 13.76 m, that of #3 coalseam was 1.91 m. The immediate roof was a coarse sandstone with an average thickness of 7.3 m, having a gray-white, coarse-grained, and massive structure. The main roof was a fine sandstone, with an average thickness of 14.7 m, having a gray-white, medium-to fine-grained, and massive structure. The immediate floor was mudstone, with an average thickness of 5 m.

3. Numerical Analysis of the Stability

3.1. Numerical Model Establishment and Simulation Plan. Based on the actual geological production conditions of the panel 8211, a plane numerical simulation calculation model was established (see Figure 2). The model was 200 m long along the x-direction. The z-axis height of the model was 108.5 m. The velocities of the horizontal and bottom boundary were set to 0. The stress applied at the upper boundary was 7.5 MPa, representing the overburden pressure. The horizontal stress was applied in the x- and y-directions of the model, and the lateral pressure coefficient was set to 1.2.

Coal and rock mass are defined as the Mohr–Coulomb model. The rock/coal mass properties required in the numerical model were obtained from the properties of an intact core by using the RocLab 10.0 software program, which is based on the generalized Hoek–Brown failure criterion (see Table 1). The simulation process was as follows: initial stress calculation balance  →  panel 8210 mining → excavation of the gob-side entry.

3.2. Analysis of Simulation Results. The distribution of vertical and horizontal displacement contours is shown in Figure 3. The obtained results are as follows:

(1) Distribution Characteristics of Vertical Displacement:
On the whole, vertical displacement of the roof coal body above the coal pillar (about 1200 mm on average) was much larger than that of the roof coal body of the entry (about 700 mm on average). From the surface of the entry to the depth of 3.5 m, the roof coal body exhibited an overall sinking trend, with an average sinking of about 800 mm, as shown in Figure 3(a). The abovementioned vertical displacement distribution characteristics can be attributed to the following reasons. (1) The thickness of the mining coal seam reached 20 m, and the entry was driven along the coal seam floor. As a result, the immediate roof of the entry was a weak and fractured coal mass with a thickness of nearly 17 m. It was prone to large-scale failure under strong mining activities, which would result in insignificant damage overall [22, 23]. (2) Compared with the panel rib, the coal pillar rib had a smaller size and weaker bearing capacity. Under the same overburden movement, its plastic failure range was larger, which in turn led to a greater vertical displacement of the roof coal body above the coal pillar.

(2) Distribution Characteristics of Horizontal Displacement:
From shallow to deep, the horizontal displacement of the panel rib gradually decreased, with the maximum displacement of 700 mm occurring in the middle of the panel rib. The coal pillar side also presented similar deformation characteristics, but the deformation value was larger than that of the panel rib side, reaching 900 mm, as shown in Figure 3(b). The reason for this phenomenon is that the bearing capacity of the coal pillar was much smaller than that of the panel rib. Therefore, it was...
necessary to increase the supporting strength of the coal pillar in the field.

The plastic zone and vertical stress distribution of the surrounding rock of the gob-side entry in panel 8211 are shown in Figure 4.

(1) Distribution Characteristics of the Plastic Zone: It can be seen from Figure 4(a) that the roof coal rock mass was in a state of large-scale shear failure, and the failure depth was about 5 m. The panel and coal pillar ribs were in shear failure models. The failure range of the panel side was about 5 m, and the coal pillar rib was in a state of plastic failure. For the shallow coal body of the roof and two sides, it was in the tensile failure model.

(2) Distribution Characteristics of the Vertical Stress: From Figure 4(b), it can be seen that the shallow coal body of the entry was in a state of stress release, with
4. Ground Control of Surrounding Rock of the Gob-Side Entry

4.1. Control Principle of Surrounding Rock of the Gob-Side Entry. Based on the actual geological production conditions and characteristics of the surrounding rock of the gob-side entry, the process of coal mass deformation and failure of the tailgate 8211 was analyzed as follows. (1) Tailgate 8211 was driven along the floor line of the coal seam, and the immediate roof was 17 m thick weak coal body with developed cracks. During entry excavation, affected by mining stress, the coal body would gradually undergo a plastic failure from shallow to deep, and consequently, the roof would sink and deform significantly. (2) Under the overlying strata movement, the coal pillar rib was all in the plastic failure state, and the bearing capacity of the coal pillar was smaller. Under strong mining stress, extrusion deformation occurred to the coal pillar rib, causing significant horizontal displacement.

Based on the above numerical analysis, to ensure the safety and stability of tailgate 8211 during its service period, the ground control should start from the following aspects [24, 25]. (1) The coal pillar was totally in a state of plastic failure; as a result, its supporting force on the roof was small and it was easily failed. Therefore, improving the support intensity of the coal pillar is the key to control the deformation of the surrounding rock of the gob-side entry. (2) The roof of the gob-side entry was composed of a weak coal body. Affected by the strong mining action induced by the entry driving and the panel retreating, the coal body surrounding the entry would be severely failed and resulted in a relatively high crushing pressure. Therefore, it was necessary to adopt support components with a larger surface area, to improve the support strength of the surrounding rock of the surface.

4.2. Control Technology for the Gob-Side Entry with Narrow Coal Pillars in Extra-Thick Coal Seams. A support scheme, including high-strength bolts, strong roof anchors, and reinforced anchors for coal pillars, was determined, as shown in Figure 5. The specific parameters were as follows.
The entry roof adopted a $20 \times 2600$ mm thread steel bolt, and the row spacing was $900 \times 900$ mm. Each row was arranged with six bolts. The bolts at the two sides were inclined 15 degrees outward, and the rest were arranged vertically. The bolt was connected by a reinforced ladder beam made of $14$ mm round steel welding. Anchor cables with a diameter of $17.8 \times 8250$ mm were selected, and the spacing between rows was $1050 \times 1800$ mm. Four anchor cables were arranged in each row. The anchor cables at the two sides were inclined 15 degrees outward, 16-channel steel was used for connection, and the middle anchor cable was vertical to the roof. A $20 \times 2600$ mm thread steel bolt was selected for the panel rib and coal pillar rib, with a row spacing of $1000 \times 900$ mm, and each row had four bolts. The anchor rod at the roof was inclined 15 degrees upward, and the anchor rod at the floor was inclined 15 degrees downward. The rest were arranged vertically on two sides and connected by steel beams welded by $12$ mm round steel. On the coal pillar rib, the prestressed anchor cables were arranged in the middle of two adjacent rows of bolts with a distance of $900$ mm. The upper anchor cables were inclined 15 degrees upward, and the bottom anchor cables were inclined 15 degrees downward.

5. Engineering Application

During the excavation of the tailgate 8211, four measuring stations were arranged in the entry with a distance of 50 m. The JSS30 A digital display convergence meter was used for measurements. The displacement of the entry roof and two sides during excavation period is shown in Figure 6. It can be seen that the surface displacement of the entry showed a changing trend of “coal pillar rib > solid coal rib > roof.” Finally, the deformation of the coal pillar was 131 mm, the deformation of the panel rib was 125 mm, and the subsidence of the roof was 99 mm. It can be seen that the deformation of the surrounding rock of tailgate 8211 was within a controllable range, which could meet the needs of...
normal panel mining. The photo of site support effect is shown in Figure 7. It should be noted that the applicability of the proposed support scheme on other coal mines needs to be studied because every coal mine may have different geological and mining conditions, which greatly affect the support parameters design. Further case studies are needed in order to deliver some general principles of support scheme design.

6. Conclusion

(1) Because the immediate roof was composed of weak coal mass with a thickness of 17 m, the roof coal mass was vulnerable to failure on a large scale with the effect of overlying strata pressure; thus, a visual subsidence of roof coal mass with a maximum convergence of 800 mm was observed.

(2) The bearing capacity of the coal pillar was significantly less than that of the panel rib, resulting in the pillar failing more easily under the ground pressure and then generating large-scale squeezing deformation along the horizontal direction.

(3) On the basis of the coal mine’s geological production conditions and the deformation and failure laws of the surrounding rock along the gob-side entry, a support scheme including an asymmetric anchor beam truss, roof angle anchor cable, and anchor cable combination structure was proposed.

It should be noted that the optimal support scheme and coal pillar size strongly depend on the geological and mining conditions. In addition, this study was only based on a specific coal mine model. Further case studies are needed in order to deliver some general principles of gob-side gate road stability design. However, the modelling procedures presented in this study are necessary in the design of yield pillars in other coal mine.

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 that they have no conflicts of interest.

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