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Article

Determination of Critical Damage Size of Inclined Waterproof Coal Pillar under Asymmetric Load

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Abstract: Quantitative determination of the critical size of inclined coal pillar in old goaf water affected area is of great significance for water damage prevention and safe mining. The critical size of inclined waterproof coal pillar is derived by using theoretical analysis, numerical calculation and field engineering practice for the stability of the waterproof coal pillar in old goaf water affected area of 1303 working face of Dananhu I mine in Xinjiang region. Firstly, a force model of inclined waterproof coal pillar was established to reveal the law that the critical size of the coal pillar increases with the increase of coal seam inclination under the action of asymmetric load. Then, numerical simulation was applied to reveal the dynamic evolution process of plastic deformation-destabilization-destruction of section coal pillar under the influence of mining and single-side water pressure, and the critical size of waterproof coal pillar in study area was determined to be 19.09m. Finally, measures such as pumping and pressure relief and slurry reinforcement were adopted to reduce the deformation rate of the roadway on side of waterproof coal pillar, which ensured the stability of the coal pillar and roadway and the safe recovery of the working face.

Keywords: inclined coal seam; asymmetric loading; waterproof coal pillar; critical size

1. Introduction

Over 60% of China's electricity is generated through the combustion of coal, making the secure and efficient extraction of coal resources vital to the nation's economic development. One key aspect of ensuring the safety of coal mining operations is the rational design and stability control of coal pillars [1–4]. Additionally, given China's complex underground hydrological conditions, with up to 25 billion tons of coal reserves affected by water, the design of many coal pillars must not only consider the safe bearing of overlying strata loads and the stability of surrounding roadways but also the prevention of water-related hazards in voids left by coal extraction [5,6]. However, coal pillars, as underground support structures, are inherently fraught with complexities due to their geological formation and the presence of numerous fractures [7–9]. Human-induced disturbances, such as mining activities, can easily weaken these structures internally [10,11]. When the load-bearing capacity of a coal pillar reaches its yield limit, catastrophic events like structural collapse are highly likely, resulting in significant economic losses and potentially even loss of life [12–14].

To address these issues, numerous scholars have conducted extensive research on waterproof coal pillars, yielding significant findings. For example, theories like load estimation methods have been used to analyze the characteristics of overlying strata loads on coal pillars, focusing on these loads as the primary source of instability and elucidating the mechanisms behind pillar failure [15,16]. For example, by drawing on the two-zone constraint theory, the water barrier coal pillar is divided into three parts of the central elastic zone and the plastic zone on both sides, and using the relevant theory of elastic mechanics, the critical size formula of the elastic zone of the coal pillar in near horizontal coal seam under the combined effect of uniform load and water pressure is derived [17–

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20]. For example, using the theory of hydrodynamics and multiple regression [21–23], we analyze the reasonable width of the water barrier coal pillar under the influence of different types of faults, and derive the formula for calculating the size of the water barrier coal pillar when the fault dip angle is large and small [24–27]. For example, neural networks and improved numerical calculation algorithms have been employed for reasonable analysis and prediction of coal pillar dimensions [28,29]. The analysis of the influence of the overburden collapse process and the development characteristics of the water-conducting fracture zone on the stability of the water barrier pillar is given by means of physically similar simulation [30–35]. In recent years, with the popular development of computer, numerical simulation has gradually become a new means to study the size of waterproof coal pillar [36–40]. Many scholars mostly use the seepage and deformation theory to establish the numerical model of solid-flow coupling in the coal seam mining process to reveal the stress evolution law of coal pillars of different widths under the influence of water and mining, which is used as the basis for the size of water barrier pillars [41–44].

Despite these advancements, current research rarely addresses the impact of coal seam inclination on the form of coal pillars or explores the stress characteristics at different locations within the pillars. This paper, therefore, focuses on the asymmetrical loading on inclined coal pillars caused by their angle of inclination. It examines the stress patterns and critical instability characteristics of the elastic barrier zone in inclined pillars, studies the dynamic evolution of plastic deformation-instability-failure of the inclined waterproof coal pillar under the influence of mining activities and unilateral water pressure, and establishes expressions for the critical failure dimensions of the inclined coal pillar under unilateral hydrostatic pressure. This provides a reference for the stability research of inclined coal pillars in similar geological conditions and offers scientific theoretical support for the prevention of disasters caused by underground space instability.

2. Mechanics Modeling

2.1. Force Analysis of Inclined Waterproof Coal Pillar

Artificial mining disturbance is the key factor to break the balance of stress between the coal pillar and its surrounding rock and lead to the formation of old goaf water. In the actual production process, with the back mining of two working faces and the gradual generation of old goaf water in the mining area, both ends of the water barrier coal pillar will be affected by the support pressure, and one side will also be affected by old goaf water, which will certainly produce stress concentration in both sides of the coal pillar. When the degree of stress concentration exceeds the bearing capacity of the coal pillar body, the internal joint fracture of the coal pillar will gradually expand, resulting in the transformation of the structure of both sides of the coal pillar from elastomeric to plastic body. When the plastic zone on both sides expands to penetration, the coal pillar will lose its load-bearing and water-insulation capacity.

In this process, as the force characteristics along the direction of roadway advancement in the coal pillar change minimally, the cross-section of a coal pillar is typically the focus of study. This reveals that the ends of a coal pillar bear more load, leading to more fracture development and a transition from an elastic to a plastic state. Closer to the middle of the pillar, the overlying strata load approaches the original rock stress, resulting in less transformation between elastic and plastic states in the coal pillar.

Therefore, in this paper, the coal pillar under the influence of water addition in the mining area is divided into the mining disturbance zone (MDZ), the elastic barrier zone(EBZ) and the water pressure affected zone(WAZ), as Fig1 shown. In this figure, G is the overburden load (vertical direction) under mining disturbance; q is the hydrostatic pressure value on one side of the coal pillar; α is these inclination angle of the coal seam; H is the height from the middle of the elastic zone of the coal pillar to the ground surface.

2.2. Determination of the Critical Length of the Elastic Barrier Zone

From the instability process of the overburden load on inclined pillars, it's inferred that the cross-sectional structure of an inclined pillar can be simplified into shorter mining-disturbed zones at both ends and a longer central elastic zone (EBZ). Since the bearing capacity of the plastic zones on both sides is compromised, the central elastic body should be the main focus of the study. Since the force direction in the elastic barrier area of the waterproof pillar is mainly concentrated in the XOY plane (Figure 1), and the force characteristics of the tangent plane perpendicular to the Z axis (parallel to the XOY plane) are similar, the force state of the coal pillarcan be simplified to a plane problem for solution [14].

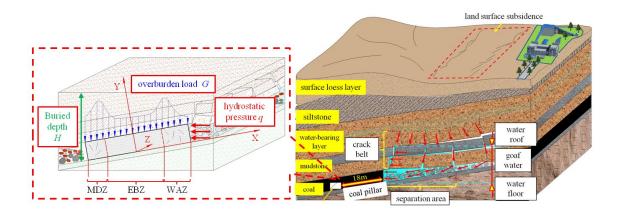
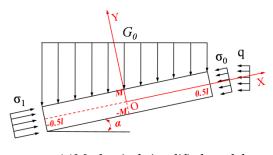
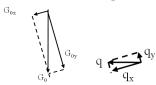


Figure 1. Overburden load of inclined waterproof coal pillar.

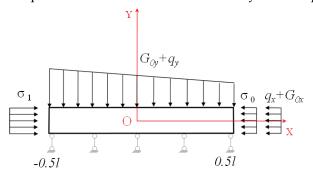
Assuming that the length of the elastic barrier zone(EBZ) is l, the overburden load of coal pillar is G_0 , and the hydrostatic pressure q is constant after crossing the water pressure influence zone, a simplified model of the elastic barrier zone mechanics can be obtained, as shown in Figure 2a.



(a)Mechanical simplified model



(b)Decomposition of overburden load *G*₀ and hydrostatic pressure *q*



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(c)Stress characteristics of Model after clockwise rotation α Angle

Figure 2. mechanical model of elastic barrier zone of inclined waterproof coal pillar.

When X-axis is the horizontal direction and Y-axis is the vertical direction, then σ_{θ} is the horizontal constraint between the coal body in the elastic zone and the water pressure influence zone; σ_1 is the horizontal constraint between the coal body in the elastic zone and the recovery disturbance zone; *q* is the hydrostatic pressure value after crossing the plastic zone of the coal pillar; α is the dip angle of the coal seam. It is known from the literature [8,14] that under the near horizontal coal seam conditions, the residual support pressure on the elastic zone of the coal pillar can be approximated as constant as $K_1 \gamma H$, where K_1 is the stress concentration coefficient of the elastic zone (generally half of the maximum stress concentration coefficient).

Through the above elaboration, we can analysis stress characteristics in the elastic barrier zone of inclined waterproof coal pillar. As shown in Figure 2b, G_0 can be decomposed into G_{0x} and G_{0y} , qcan be decomposed into q_x and q_y . So stress of G_{0x} and q_x is horizon vector, stress of G_{0y} and q_y is vertical vector. Thus,

$$G_{0} = -K_{1}\gamma(H - x\sin\alpha)$$
(1)
$$G_{0y} = -K_{1}\gamma(H - x\sin\alpha)\cos\alpha$$
(2)
$$G_{0x} = -K_{1}\gamma(H - x\sin\alpha)\sin\alpha$$
(3)
$$q_{x} = -|q|\cos\alpha$$
(4)
$$q_{y} = |q|\sin\alpha$$
(5)

Because the vertical stress of EBZ does not change along with the change of Y axis direction stress, G₀ is constant.

Thus, using semi-inverse solution of elastic mechanicsto can solve this problem.

In Figure 2a:

To simplify the calculation, definition:
$$A = \frac{K_1 \gamma \sin 2\alpha}{2} x + (|q| \sin \alpha - K_1 \gamma H \cos \alpha)$$

$$A = \frac{K_1 \gamma \sin 2\alpha}{2}, \quad B = |q| \sin \alpha - K_1 \gamma H \cos \alpha$$
Therefore, according to the equilibrium differential equation
$$\sigma_y = \frac{\partial^2 f(\mathbf{x}, \mathbf{y})}{\partial x^2}.$$

By integrating Equation (6), the stress function can be obtained □

$$f(x,y) = \int \sigma_y dy = \frac{A}{6}x^3 + \frac{B}{2}x^2 + f_1(y)x + f_2(y)$$
 (7)

We can find that $f_1(y)$ and $f_2(y)$ are functions of the unknown parameter y. In order for the stress function f(x,y) to hold, the compatibility equation should be satisfied:

$$\frac{\partial^4 f(x,y)}{\partial x^4} + 2 \frac{\partial^4 f(x,y)}{\partial x^2 \partial y^2} + \frac{\partial^4 f(x,y)}{\partial y^4} = 0$$
(8)

Eq. (7) is substituted into Eq. (8) to obtain:

$$x\frac{\partial^4 f_1(y)}{\partial y^4} + \frac{\partial^4 f_2(y)}{\partial y^4} = 0$$
 (9)

Since $x \in [-0.5l, 0.5l]$, taking different values of x and substituting it into Eq. (4), simplifying it gives:

$$\frac{\partial^4 f_1(y)}{\partial y^4} = 0 \qquad \frac{\partial^4 f_2(y)}{\partial y^4} = 0$$

Therefore, the expressions of $f_1(y)$ and $f_2(y)$ can be obtained:

$$f_1(y) = C_1 y^3 + C_2 y^2 + C_3 y + C_4$$
 (10)

$$f_2(y) = D_1 y^3 + D_2 y^2 + D_3 y + D_4$$
 (11)

And substituting Eq. (10) and Eq. (11) into Eq. (7), it is apparent that:

This leads to
$$\sigma_{x} = \frac{\partial^{2} f(x, y)}{\partial y^{2}} = 6C_{1}xy + 2C_{2}x + 6D_{1}y + 2D_{2}$$

$$\sigma_{y} = \frac{\partial^{2} f(x, y)}{\partial x^{2}} = Ax + B$$

$$\tau_{xy} = -\frac{\partial^{2} f(x, y)}{\partial x \partial y} = -(3C_{1}y^{2} + 2C_{2}y + C_{3})$$
(15)

As the elastic zone of coal pillar is in critical equilibrium, and there is a large interlayer friction between coal and rock seams, so \square

$$\sigma_1 = -(\sigma_0 + q_x + G_{0x})$$
 (16)

Based on this simplified calculation, the symmetry of the mechanical model is therefore considered first. As shown in Figure 2c, since the bottom of the coal pillar can be regarded as a fixed end constraint, the load on the elastic barrier zone of the water barrier coal pillar is symmetric along the XOZ side, so it should satisfy $\sigma_x(-y) = \sigma_x(y)$, $\sigma_y(-y) = \sigma_y(y)$, $-\tau_{xy}(y) = \tau_{xy}(-y)$. Substituting into equations (13)-(15) yields $C_1 = D_1 = C_3 = 0$.

At this point, considering the boundary conditions, as shown in Figure 2c, at the boundary x = -0.5l, it is obtained that

$$\sigma_x(-0.5l) = -C_2l + 2D_2 = -(\sigma_0 + q_x + G_{0x})$$
 (17)

At the boundary x = 0.5l, it follows that

$$\sigma_x(0.5l) = C_2l + 2D_2 = \sigma_0 + q_x + G_{0x}$$
 (18)

The union of (17) and (18) is solved by $D_2 = 0$, $C_2 = \frac{\sigma_0 + q_x + G_{0x}}{l}$, and combined with equations (13)-(15) we get:

$$\sigma_{x} = 2 \frac{\sigma_{0} + q_{x} + G_{0x}}{l} x$$

$$\sigma_{y} = G_{0y} + q_{y} \quad (20)$$

$$\tau_{xy} = -2 \frac{\sigma_{0} + q_{x} + G_{0x}}{l} y \quad (21)$$

in the formula, $x \in [-0.5l, 0.5l]$, $y \in [-0.5M, 0.5M]$.

Since the force in the horizontal direction of the coal pillar (with the coordinate axis as the reference system) is symmetrical along the *Y*-axis and the object has been assumed to be a uniform medium, the horizontal combined stress at the point *O* at its form center should be 0 and should only be subjected to the pressure in the vertical direction. At this time, x=0 and y=0 are substituted into equations (19)-(21), and σ_x and τ_{xy} are both 0, thus verifying the accuracy and reasonableness of the above equation.

The magnitude of the principal stress at any point within the elastic barrier of the coal pillar can be obtained by combining the elastodynamic principal stress relationship equation.

At this point, considering the underground engineering problem, the damage of the surrounding rock is suitable for analysis using the Mohr–Coulomb criterion, so the expression for the stress at it critical damage is introduced.

$$\sigma_1 = \frac{1 + \sin \varphi}{1 - \sin \varphi} \sigma_3 + \frac{2c \cos \varphi}{1 - \sin \varphi}$$
 (22)

Where, c is the cohesion of coal body and φ is the friction angle within the coal body.

At this time, the critical size of the elastic barrier zone can be obtained by substituting any point σ_x , σ_y , τ_{xy} , at the boundary of the waterproof pillar into equation (22) and connecting vertical (19)-

(21).To simplify the calculation, The boundary point (0,0.5M) is substituted as shown in Figure 2c, x=0 and y=0.5M (M is the thickness of the coal seam).well sorted:

$$l = \frac{2M(K_1 \gamma H \sin \alpha + |q|\cos \alpha + |\sigma_0|)}{\sqrt{\left[4c\cos \varphi + 2(|q|\sin \alpha - K_1 \gamma H \cos \alpha)\sin \varphi\right]^2 - (|q|\sin \alpha - K_1 \gamma H \cos \alpha)^2}}$$
(23)

Where, according to the Cullen Moore criterion, it is obtained that

$$\sigma_0 = \frac{1 - \sin \varphi}{1 + \sin \varphi} K \gamma H - \frac{2c \cos \varphi}{1 + \sin \varphi}$$
 (24)

Equation (23) is the expression of the critical size of the elastic barrier zone of the inclined coal pillar (inclination angle α) under the influence of water pressure q and recovery disturbance (to reflect the directionality of the force more intuitively, the force in the negative direction with the coordinate axis is therefore added to the absolute value).

And when the coal seam dip angle is 0° , it is known that $sin\alpha = 0$ and $cos\alpha = 1$. The critical size l of the barrier zone in the middle of the waterproof coal pillar of the horizontal coal seam can be obtained by bringing in equation (23):

$$l' = \frac{2M(|q| + |\sigma_0|)}{\sqrt{|(4c\cos\varphi - 2K_1\gamma H\sin\varphi)^2 - (K_1\gamma H)^2|}}$$
(25)

2.3. Calculation of the Critical Length of the Elastic Barrier

It can be deduced that based on the characteristics of the overburden load on the inclined waterproof coal pillar, combined with the functional relationships between various physical parameters and the above derivation process, it is apparent that:

- (1) The numerator of equation (22), $K_1\gamma H sin\alpha + |q|cos\alpha + |\sigma_0|$, is exactly the main stress σ_x applied to both sides of the boundary of the elastic barrier zone of the coal pillar, and the denominator, $|q|sin\alpha K_1\gamma H cos\alpha$, is the main stress σ_y applied to the upper boundary of the elastic zone of the coal pillar, while the rest are the mechanical parameters of the coal pillar and its dip angle. It can be seen that the critical length of the central barrier zone is closely related to the principal stress σ_x in the inclination of the elastic zone of the coal pillar, the principal stress σ_y in the vertical direction and the inclination angle of the coal seam.
- (2) By analysing equations (19)-(21) and comparing them with equation (23), we can see that when the dip angle of the seam is 0°, the forces at the bottom angle of the pillar are equal and maximum, and damage can occur at any position at the bottom of the pillar. When the inclination angle α is not 0°, the internal stresses in the pillar are greater at the corners, but the stresses at the lowest horizontal part of the pillar (e.g., the lower left corner of the pillar in Figure 2a) are the greatest, and σ_x , σ_y and τ_{xy} are the maximum values at this time. Therefore, when the load on the inclined coal pillar exceeds its limit, the lowest horizontal bottom angle of the pillar should be the most vulnerable to damage.
- (3) In the range of inclination angle $\alpha \in [0^{\circ}, 90^{\circ}]$, as the inclination angle increases, the load given to the coal pillar by the residual support pressure G_{θ} along the inclination of the seam gradually increases, while the load perpendicular to the coal pillar gradually decreases, i.e., σ_x is proportional to the inclination angle α , and σ_y is inversely proportional to the inclination angle α . Combined with equation (23), we can obtain that the critical size of the elastic barrier zone of the water barrier coal pillar presents a characteristic proportional to the inclination angle of the coal seam.
- (4) The destructive effect of water on the coal pillar is mainly reflected in two aspects. On the one hand is the pressure effect on the coal pillar, as can be seen from equations (23) -(25), with the increase of water pressure q, mainly reflected in the influence of the main stress σ_x on the tendency of the coal pillar, and the greater the water pressure, the greater the critical length. The other side is that the physical and mechanical properties of the coal rock body under water immersion will be significantly reduced, which will also affect the determination of the critical length of the elastic barrier zone of the water barrier coal pillar.

Determining the length of mining disturbance zone(MDZ) and water pressure affected zone(WIZ) is l_1 and l_0 . There are extremely well established results to draw on for the study of the critical size of the coal pillars disturbed by mining at the working face. The literature [7,8] gives an expression for the critical size of the water barrier coal pillar mining disturbance zone (MDZ) by applying the Cullen-Moore criterion and has been tested in a wide range of applications.

$$l_{1} = \frac{M}{2f\lambda} \ln \left| \frac{fK_{1}\gamma H + c}{(\lambda - 1)fc\cot\varphi + c} \right|$$
 (26)

Where, f is the friction factor between the top and bottom of the coal seam, generally f=tan ϕ /4; λ is the lateral pressure coefficient, generally taking the value of 2.5.

As for the formula for calculating the length of the water pressure affected zone in inclined coal seams, the literature [22] combined theoretical and empirical formulas and gave its ratio to the length of the recovery disturbance zone.

$$\frac{l_1}{l_0} = \frac{\cos(\delta + \alpha)}{\cos(\delta - \alpha)} \tag{27}$$

Where, α is the coal pillar inclination angle; δ is the rock seam movement angle.

Therefore, the final combined equations (23)-(27) can be obtained as the critical length of water barrier coal pillar in inclined coal seam:

$$= \left[1 + \frac{\cos(\delta - \alpha)}{\cos(\delta + \alpha)}\right] \frac{M}{2f\lambda} \ln \left| \frac{fK_1\gamma H + c}{(\lambda - 1)fc\cot\varphi + c} \right| + \frac{2M(K_1\gamma H\sin\alpha + |q|\cos\alpha + |\sigma_0|)}{\sqrt{\left[4c\cos\varphi + 2(|q|\sin\alpha - K_1\gamma H\cos\alpha)\sin\varphi\right]^2 - (|q|\sin\alpha - K_1\gamma H\cos\alpha)^2}}$$
(28)

3. Application of Critical Size Calculation for Water Barrier Coal Pillars

3.1. Project Overview of the Study Area

The 18m coal pillar is left between the empty area of 1301 working face and the auxiliary transport roadway of the current 1303 working face in Dananhu I mine. As a result of the fissures generated by the mining disturbance during the mining process of the upper section (1301 working face), part of the coal pillar in the section (745-1114 m section from the return wind tunnel) is affected by the water accumulation in the mining void area (Figure 3), whose accumulation elevation is +209m and is non-flow dynamic, and the maximum hydrostatic pressure is 0.26MPa. Therefore, it is important to determine the limit of stability of the water-insulated coal pillar, so as to predict whether the existing pillar can ensure the safe recovery of the working face under the influence of water.

Through multiple means, including field collection, coal rock sampling experiments and physically similar simulations, the relevant parameters required for the waterproof coal pillar calculation were finally obtained (as shown in Table 1)

	•	•	•		
	Plastic zone				
Elastic zone stress	tress	Dip angle of coal seam	Coal seam	Rock volume force γ(KN/m³)	Buried depth <i>H</i> (m)
concentration factor	concentration		thickness		
\mathbf{K}_1	factor	(°)	(m)		
	K				
2	4	12	6.5	24.5	255.5
			Maximum	Strata	Friction
Internal friction angle	Cohesion $c(MPa)$	Poisson ratio	hydrostatic	movement	factor
$arphi(^\circ)$		v_0	pressure	angle δ	
			q(MPa)	(°)	ſ
27.4	1.92	0.36	0.26	50	0.129

Table 1. Physical and mechanical parameters of coal pillar in Dananhu I mine.

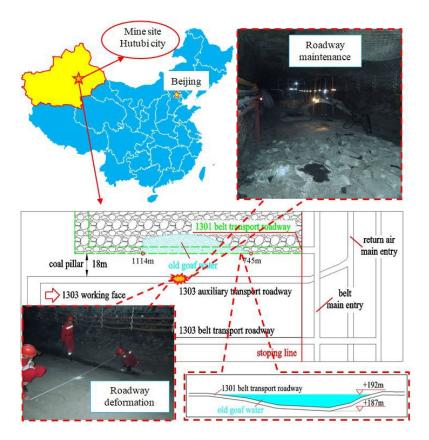


Figure 3. 1303 Working face mining schematic diagram.

3.2. Calculation of the Critical Size of the Waterproof Coal Pillar in the Study Area

Substituting the above parameters into equation (26), we can get the critical size l_1 of 2.95m for the disturbance zone of waterproof coal pillar retrieval. and then from equation (27), we can get the critical size l_0 of 4.95m for the water pressure influence zone.

Substituting the above dimensional results and related parameters into equation (24), the stress σ_0 at the junction of the plastic and elastic zones can be obtained as 6.921 MPa.

Finally, by substituting all the relevant parameters into equation (23), the critical size 1 of the elastic barrier zone of the coal pillar in the limit state:

$$\begin{split} I &= \frac{2M(K_1\gamma H \sin\alpha + |q|\cos\alpha + |\sigma_0|)}{\sqrt{\left[4c\cos\varphi + 2(|q|\sin\alpha - K_1\gamma H \cos\alpha)\sin\varphi\right]^2 - (|q|\sin\alpha - K_1\gamma H \cos\alpha)^2|}} \\ &= \frac{2\times6.5\times(2\times\frac{24.5\times255.5}{10^3}\times0.208 + 0.26\times0.978 + 6.921)}{\sqrt{\left[4\times1.92\times0.8878 + 2\times(0.26\times0.208 - 2\times\frac{24.5\times255.5}{10^3}\times0.978)\times0.46\right]^2 - (0.26\times0.208 - 2\times\frac{24.5\times255.5}{10^3}\times0.978)^2}} = 11.19\text{m} \end{split}$$

Therefore, in the limit state, the critical size of the inclined state waterproof coal pillar is: $L = l_1 + l + l_0 = 2.95 + 11.19 + 4.95 = 19.09 \text{m}$

And when the coal seam inclination angle α is assumed to be 0°, the critical dimensions of the elastic barrier zone l' and the water pressure influence zone l'0 of the horizontal state coal pillar can be obtained by substituting equations (25) and (27) as follows:

$$l' = \frac{2M(|q| + |\sigma_0|)}{\sqrt{(4c\cos\varphi - 2K_1\gamma H\sin\varphi)^2 - (K_1\gamma H)^2|}}$$

$$= \frac{2\times 6.5 \times (0.26 + 6.921)}{\sqrt{(4\times 1.92 \times 0.8878 - 2\times 2\times \frac{24.5 \times 255.5}{10^3} \times 0.46)^2 - (2\times \frac{24.5 \times 255.5}{10^3})^2|}} = 7.76m$$

$$l'_0 = \frac{\cos(\delta - 0^\circ)}{\cos(\delta + 0^\circ)} = l_1 = 2.95m$$

Therefore, taking the geological conditions of Dananhu I mine as the benchmark (excluding the coal seam inclination), the theoretical limit size is 13.66m when the coal seam inclination angle α is assumed to be 0° to leave the water barrier coal pillar; with the coal seam inclination angle α of 12° to leave the water barrier coal pillar, the width of coal pillar is 19.09m, an increase of 29%. The coal seam inclination angle has an extremely important influence on the retention of coal pillars. The width of the coal pillar left at the project site is 18m, which is close to the limit size, so it is necessary to take corresponding protective measures for the coal pillar and the surrounding roadway.

4. Elastic-Plastic Evolution of the Waterproof Coal Pillar in the Study Area

4.1. Numerical Modeling of the Study Area

Flac3D was used to study the elastic-plastic evolution process of the inclined coal pillar under the action of asymmetric load. Based on the geological conditions of the Dananhu I mine, a numerical calculation model was established, as shown in Figure 4. Based on the field exploration and rock mechanics experiments, the relevant mechanical parameters were optimised and assigned, and the physical and mechanical parameters of each rock seam are shown in Table 2.

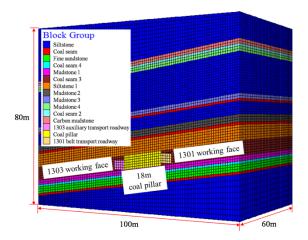


Figure 4. Numerical calculation model.

Table 2. Physical and mechanical parameters of the layers of Dananhu I Mine.

layer	compress ve strength (MPa)	i rock density (kg/m³)	bulk modulus (GPa)	shear modulus (GPa)	internal friction angle $\varphi(^{\circ})$	cohesion <i>c</i> (MPa)	poisson ratio vo
Siltstone	1.46	2360	7.8	2.8	29.9	5.35	0.29
Fine sandstone	1.33	2259	6.8	2.4	30	4.2	0.34
Siltstone	1.46	2360	7.8	2.8	29.9	5.35	0.29
Coal seam	0.48	1370	4.4	0.5	27.4	1.92	0.36
Mudstone	1.37	2397	6.07	1.37	29	3.07	0.28
Carbon mudstone	1.35	2400	6.8	1.9	28.6	4.07	0.28

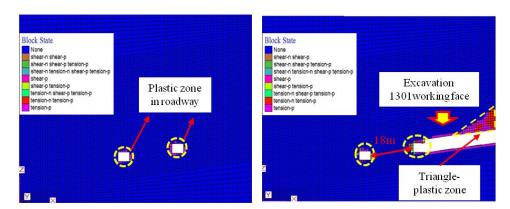
4.2. Fracture Evolution and Plastic Zone Formation Process in the Coal Pillar

Under such conditions, internal fissures in the coal pillar will gradually develop and expand, resulting in a reduction in the overall strength of the pillar and local plastic deformation. In the case of inclined coal pillar, due to the influence of its inclination angle, the force in the coal pillar will not

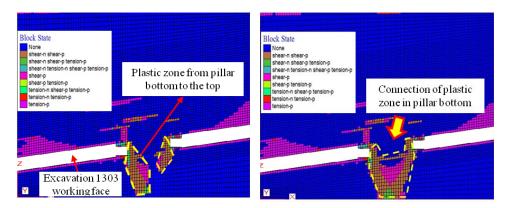
be symmetrically distributed along its center line, which will certainly lead to stress concentration in some areas of the coal pillar and eventually the strength deterioration will occur first in the stress concentration area.

Figure 5 reflects the evolution of the transformation of the elastic and plastic zones within the inclined waterproof coal pillar during the mining process, as can be seen from the figure:

- (1) From Figure 5a to Figure 5b, it can be seen that with the retrieval of the upper section working face (1301 working face) and the excavation of the roadway in this section, the internal stress equilibrium of the coal pillar is broken, and the roof plate of the 01 working face mining goaf area is broken, forming residual supporting pressure along the coal seam tendency, and finally a triangular plastic zone appears in the upper right corner of Figure 5b. At this time, the coal pillar was in a unilateral loaded state, and the fissures in the surrounding rock of the roadway on the loaded side extended and expanded, so the loose range of the surrounding rock of the roadway on the side of the coal pillar in the mining void area was about 1m larger than that of the roadway on the other side of the coal pillar, but at this time, most areas of the coal pillar were still elastic zones, indicating that the internal fissures of the coal pillar were less developed and had good stability under the unilateral loaded state.
- (2) The change process from Figure 5c to Figure 5d shows that when the working face of this section (03 working face) is retrieved, the coal pillar is in the loaded state on both sides, at this time, the coal pillar is changed from unilateral pressure to pressure on both sides, and the range of its plastic zone on both sides is further increased. However, due to the influence of the coal seam inclination angle, the loading on both sides of the coal pillar is not symmetrical, which leads to the lower left corner of the coal pillar becoming the residual support pressure concentration area, at this time, the residual support pressure not only affects the stability of the coal pillar, but also by the transfer along the coal seam floor, plus the coal seam floor is mudstone with low strength, so it leads to the obvious development of fissures in the lower left corner of the coal pillar and the formation of a large plastic zone, which eventually gradually This is also consistent with the theoretical calculation of the location of the first damage to the coal pillar. This is also consistent with the theoretical calculation where the coal pillar was first damaged. During the whole process of fracture evolution and plastic zone formation, the fractures in the surrounding rock of the roadway on both sides of the coal pillar also expanded to a certain extent, which led to a significant increase in the loosening of the roadway, indicating that the stability of the roadway was also significantly reduced under the influence of mining disturbance.
- (3) In the numerical simulation, the plastic zone of the coal pillar base plate is penetrated, so the plastic penetration zone will easily become the old goaf water seepage channel and eventually induce water damage. Therefore, it is necessary to carry out corresponding old goaf water prevention and coal pillar reinforcement measures in advance to ensure the safe recovery of the working face.



(a) excavation roadway and (b) excavation 1301 working face



(c) model operation midway after excavation 1303working face (d) model operation completed

Figure 5. Evolution process of elastic-plastic zone in waterproof coal pillar after mining influence.

5. Engineering Practice

In view of the weakening of the coal pillar by the old goaf water and the characteristic that the bottom angle of the coal pillar is most easily damaged due to the inclination angle, the secondary reinforcement of the water barrier coal pillar is carried out by means of pumping and pressure relief and grouting reinforcement in the area of serious deformation of the coal pillar.

5.1. Pumping and Pressure Relief

In order to achieve the pumping and pressure relief, four pumping holes were constructed from 11 September in the water pressure affected area between 745m and 1114m in the 1303 auxiliary transport roadway (the horizontal elevation of the roadway in this area is between 183-202m) to carry out the water pumping and pressure relief works in the mining area. The locations of the four pumping holes are shown in Table 3.

hole position near return air pumping holes horizontal elevation drilling parameters roadway ZK-321 201m 818m drilling depth is 19m, ZK-322 192m 891m drilling diameter is ZK-323 187m 964m 52mm ZK-324 199m 1037m

Table 3. Layout position of pumping hole and drilling parameters in gob water affected area.

The water pressure of the old goaf water measured by the borehole during the pumping process is shown in Figure 6. As can be seen from the graph, the rate of decrease in water pressure shows a gradual decrease in the process of decreasing the volume of water in the old goaf water. This is due to the fact that as the pumping works progress, the water pressure measured in each borehole also approaches 0 in turn, and therefore when a borehole is completed pumping and sealed (1m concrete seal), the total daily volume of water pumped out will decrease. Ultimately, as at October 2016, a total of 12,222m³ of water had been pumped from the Dananhu I Mine and the water pressure measured at the completion of the works was 0.028MPa, indicating that this area contains only a very small amount of old goaf water.

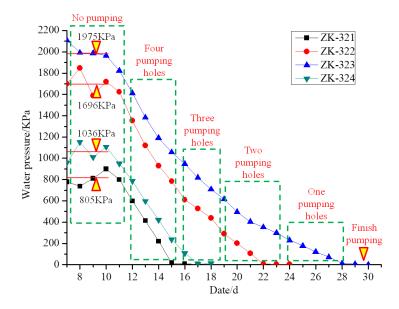
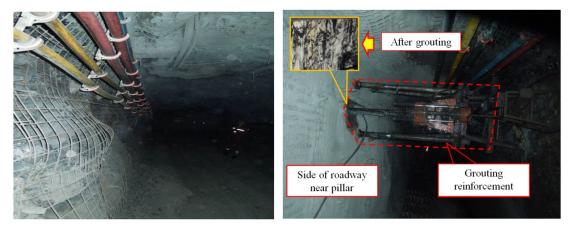


Figure 6. Water pressure measurement curve after pumping.

5.2. Grouting Reinforcement

Due to the obvious bottom drum and flaky gang phenomenon in the 1303 auxiliary transport roadway in the old goaf water affected area during the workface retrieval process, especially the bottom plate drums out obviously near the coal pillar side, as shown in Figure 7a. Therefore, in order to reduce the further expansion of the plastic deformation area of the water barrier coal pillar, polyimine gum grease was injected into the serious plastic deformation area on the coal pillar side of the tunnel, so as to achieve the purpose of closing the fissures in the surrounding rock and ensuring the structural stability of the coal pillar, as shown in Figure 7b.

As the bottom corner of the coal pillar is the starting point of plastic deformation expansion, a 42mm diameter borehole with a depth of 5m was constructed along a 45° downward direction from the coal pillar gang to inject marisan. The polyimine gum grease resin was mixed with the catalyst at a ratio of 1:1 and then injected into the coal pillar using a special polyimine gum grease grouting pump at a pressure of 8.0-18.0MPa.



(a) Tunnel bottom drum (b) Grouting reinforcement by polyimine gum grease

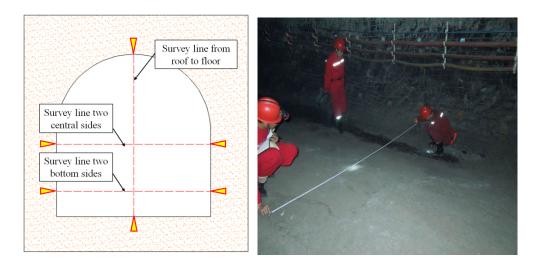
Figure 7. Grouting reinforcement by polyimine gum grease.

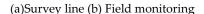
5.3. Effectiveness Test

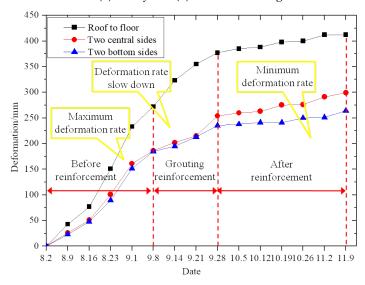
Figure 8 reveals the deformation pattern of the surrounding rock of the coal pillar in the old goaf water affected area after taking corresponding measures. The deformation of the top and bottom

plates of the roadway near the coal pillar and the two sides of the roadway were analysed in chronological order, and the overall deformation rate of the coal pillar side of the roadway showed a decreasing trend. The second stage was the period of pumping and pressure relief and polyimine gum grease injection (8 September-28 September), when the three deformation rates of the roadway were significantly reduced, including 5.6mm/d for the top and bottom plates, 3.5mm/d for the central two sides and 2.5mm/d for the lower two sides; the last stage was the post-reinforcement monitoring period (29 September-9 November), when the three deformation rates of the roadway were significantly reduced. The last stage was the post-reinforcement monitoring period (29 September-9 November), when the deformation in all three directions of the roadway was smaller, with an average of 0.84mm/d.

Through the above analysis, it can be concluded that the old void water control and coal pillar reinforcement measures have effectively curbed the expansion of coal pillar and roadway surrounding rock deformation, thus ensuring the safe recovery of the working face.







(c) Deformation analysis of roadway near the coal pillar

Figure 8. Effectiveness test of roadway near the coal pillar.

6. Conclusions

Xinjiang, as a key energy base in China's "Belt and Road" development plan, has many coal seams that are inclined and often affected by groundwater. If mines fail to control the stability of

inclined waterproof coal pillars, serious safety accidents are highly likely. Therefore, this paper conducts a mechanical analysis and numerical simulation study on the stability characteristics of the inclined waterproof coal pillar in Xinjiang mining areas under the influence of old goaf water. The findings are crucial for the safe extraction of coal resources and the promotion of stable energy supply. The specific research outcomes include:

- (1) Establishing a mechanical model for the elastic barrier zone of the inclined waterproof coal pillar, and deriving expressions for the critical dimension under the asymmetric loading. The study indicates that the limit dimension of inclined section coal pillar is significantly influenced by the angle of inclination. In scenarios where other geological factors remain constant, the critical dimensions of the elastic barrier zone in inclined section coal pillar show a direct proportional relationship with the coal seam inclination. Taking the geological conditions of the Dananhu I mine as an example, the critical width of the waterproof coal pillar at a 12° inclination increases by 30% compared to when the inclination is 0° .
- (2) Through numerical simulations, the evolution of the elastic-plastic zones in coal pillars in inclined sections under multiple mining disturbances is analyzed. Combined with mechanical analysis, it is found that the lowest bottom angle of the coal pillar is the first to experience failure. Numerical simulation experiments show that fractures first develop and expand at the lowest point of the pillar, leading to widespread plastic yielding, after which the plastic zone at this bottom corner extends along the floor of the coal seam towards the other bottom corner, eventually creating a channel for old goaf water seepage through the interconnected plastic zone at both bottom corners of the pillar.
- (3) The study also analyzed how stress concentration in the bottom corner regions of the coal pillar, caused by overburden, is key to triggering the deterioration of pillar structural stability. Based on this, methods such as pumping water to relieve pressure and grouting to reinforce the coal pillar are used to enhance the stability of the inclined waterproof coal pillar. Field practice has shown that these measures significantly reduce the deformation rate of the coal pillar side roadway, indicating that when the coal pillar is at their critical dimensions, appropriate old goaf water prevention and stability reinforcement measures can effectively restrain the deformation of the coal pillar and surrounding rock in roadways, thus ensuring the safe extraction of the workface.

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