

Article

Propagation Law of Hydraulic Fracture across the Coal-Rock Interface under the Co-Effect of Natural Fractures and Tectonic Stress

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Abstract: Indirect fracturing from roof rock to coal using a horizontal well is a new and promising technology for coalbed methane surface exploitation in soft and low-permeability coal seams. In order to study the propagation law of hydraulic fracture across the coal-rock interface, a pore pressure cohesive element is used to establish a numerical model for indirect fracturing. Combined with practical engineering in 3# coal seam in Xinjing mine in China, the propagation behavior of hydraulic fracture across the coal-rock interface was researched, and the range of horizontal well position for indirect fracturing was determined. The results show that: (1) the pore pressure cohesive element can be used to simulate accurately the interaction between hydraulic fracture and natural fracture, and the propagation of hydraulic fracture across the coal-rock interface. (2) As the vertical distance between the horizontal well and coal-rock interface decreases, the breakdown pressure of perforation decreases; while the injection pressure increases when the hydraulic fracture crosses the coal-rock interface. (3) For the indirect fracturing engineering in 3# coal seam in Xinjing mine, the vertical distance between the horizontal well and coal-rock interface should not be larger than 2.0m to make the hydraulic fracture propagate into the coal seams.

Keywords: Pore pressure cohesive element; Indirect fracturing; Natural fracture; Tectonic stress; Horizontal well position

1. Introduction

Coalbed methane (CBM) resource in China at buried depths shallower than 2000m were approximately 36.81 trillion m³ [1], while most coal seams in mining area are characterized by low saturation, low permeability, low reservoir pressure and high metamorphism [2, 3], resulting in the difficulty in the efficient and industrial development of CBM. In addition, the main coal-bearing basins in China have experienced long-term intense squeezing, shearing and deformation [4-6], which may result in the problems such as hole collapse, stuck drilling and plugging when directly fracturing the coal seam. Therefore, the indirect fracturing technology has been widely used in CBM extraction in recent years. The core idea of this technology is that the horizontal well is drilled in the roof or the floor, and the hydraulic fracture initiating in the roof or floor propagates into the coal seam. To date, the indirect fracturing technology has been successfully applied in Luling mine of Huaibei and in Zhaozhuang mine of Jincheng in China [7].

The concept of indirect fracturing for CBM extraction was firstly proposed by Olsen et al. [8-9] at the beginning of this century, and many scholars have done lots of work aimed at the propagation law of hydraulic fracture across the coal-rock interface. Tan et al. [10] experimentally studied the fracture propagation behaviour in tight sandstone-coal interbedded formations. Jiang et al [11] experimentally studied the effect of interfacial friction and in-situ stress difference on the propagation law of hydraulic fracture across the coal-rock interface. Liu et al [12] experimentally studied the effect of injection rate of the fracturing fluid on the dynamic propagation of hydraulic fracture across the

coal-rock interface. Wan et al. [13] studied transition zone's influence on the fracture vertical propagation behaviour for coal measure strata. Then the effect of fracture initiation position and fluid viscosity on the fracture propagation in multi-layered coal strata were experimentally investigated [14]. He et al. [15] found that compared with hydraulic fracturing, the stress difference between the vertical stress and the minimum horizontal principal stress for cracks to penetrate through the coal - rock interface is larger in supercritical CO₂ fracturing.

As for the simulation work, Zhao et al. [16] established a composite criterion to predict subsequent intersection behavior between a hydraulic fracture and a natural fracture. Huang et al. [17-18] using the block discrete element method to explore the influence of engineering parameters on the behaviour process of hydraulic fractures penetrating bedding planes. Li et al. [19] studied the propagation law of hydraulic fracture across the coal-rock interface based on Plasticity Fracture-Seepage (PF-S) model, and found that the in-situ stress and the interfacial shear strength are the dominant factor controlling the propagation of hydraulic fracture across the coal-rock interface. Escobar et al. [20] studied the effect of stress interference on the penetration ability of multiple fractures using XFEM method, and the results showed that stress shadowing facilitate the propagation of hydraulic fracture from shale into roof rock.

To sum up, the current researches about the indirect fracturing for CBM extraction using horizontal well mainly focus on the in-situ stress, rock mechanical parameters, interfacial strength, and fracturing operation parameters such as injection rate, interval distance between fracturing stages and so on. However, it has little consideration on the natural fractures, and the considered in-situ stresses are restricted to the conventional normal faulting stress regime, i.e. the vertical in-situ stress is the largest stress. While the strike-slip faulting stress regime in the Qinshui basin is widely distributed, i.e. the horizontal in-situ stress is larger than the vertical in-situ stress [21-22]. In addition, it is well known that the coal measure strata contains many natural fractures. Therefore, it is of practical importance to study the propagation law of hydraulic fracture across the coal-rock interface under the co-effect of natural fractures and tectonic stress.

In this paper, the applicability of the pore pressure cohesive element to the simulation of the interaction between hydraulic fracture and natural fracture and the propagation of hydraulic fracture across the coal-rock interface is validated. Then, considering the natural fracture distribution and tectonic stress field of 3# coal seams of Xinjing mine in Shanxi province in China, we established the numerical model for indirect fracturing from the roof to coal, using which the propagation law of hydraulic fracture across the coal-rock interface was studied, and the required location of the horizontal well in roof for the hydraulic fracture propagating into the coal from the roof was determined. This research can provide theoretical basis for the CBM extraction in natural fractured coal seam with tectonic stress field using indirect fracturing.

2. Cohesive element model for hydraulic fracturing

As a special finite element, the cohesive element has been widely used to simulate the fracture propagation. In terms of hydraulic fracturing simulation, the pore pressure cohesive element with zero thickness is used to pre-define the propagation path of hydraulic fracture. As shown in Fig. 1, the fracture process zone (FPZ, i.e. the unbroken cohesive zone) at the tip of hydraulic fracture is simulated using a partially damaged cohesive element with non-zero traction T , while the visible fracture (i.e. the broken cohesive zone) is simulated using the completely damaged cohesive element with the traction T of zero. The mathematic crack tip refers to the point which is yet to separate, the cohesive crack corresponds to the damage initiation point where the traction T reaches the tensile strength T_{\max} and the separation δ between the top and bottom surface of the cohesive element reaches the critical value δ_0 , and the material crack tip is the completely damaged point where the δ reaches the critical value δ_f and the cohesive strength just vanishes [23-25].

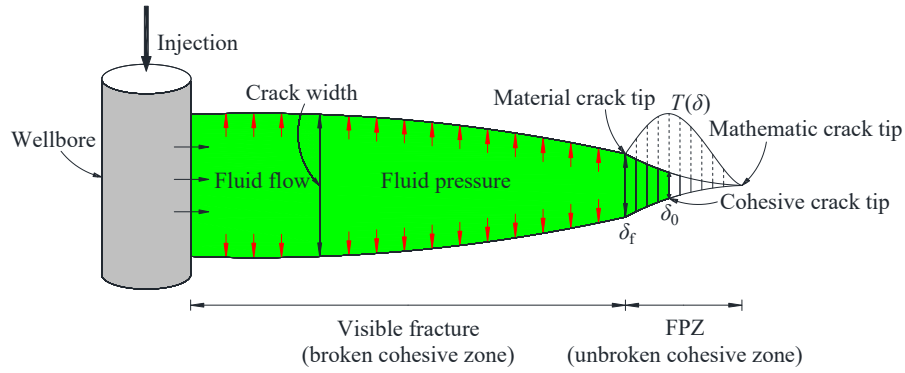


Figure 1. Cohesive element model for hydraulic fracturing.

2.1. Equations governing the solid deformation

The linear elastic constitutive relation is adopted in to describe the coal and rock deformation. Specifically, \mathbf{T} represents the nominal traction vector subjected to the top and bottom surfaces of the cohesive element. For 2D numerical model, \mathbf{T} consists two components, i.e. the one perpendicular to the surface of cohesive element and the one parallel to the surface of cohesive element, which are denoted by T_n and T_s , respectively. The corresponding separations between the top and bottom surfaces of cohesive element are denoted by δ_n and δ_s , respectively. The initial constitutive thickness of cohesive element is denoted by h_0 , and then the nominal normal strain and shear strain can be defined as

$$\begin{aligned}\varepsilon_n &= \frac{\delta_n}{h_0} \\ \varepsilon_s &= \frac{\delta_s}{h_0}\end{aligned}\quad (1)$$

Correspondingly, the linear elastic constitutive relation can be written as

$$\mathbf{T} = \begin{Bmatrix} T_n \\ T_s \end{Bmatrix} = \begin{bmatrix} E_{nn} & E_{ns} \\ E_{ns} & E_{ss} \end{bmatrix} \begin{Bmatrix} \varepsilon_n \\ \varepsilon_s \end{Bmatrix} = \mathbf{E}\boldsymbol{\varepsilon}\quad (2)$$

As the default value of h_0 , 1.0 is used in following simulation, so E in Equation. (2) is the stiffness matrix, and T_n and T_s are the normal stress t_n and shear stress t_s , respectively.

2.2. Equations governing the solid deformation

The irreversible bilinear traction-separation law [26] is adopted to simulate the damage process of cohesive element, as shown in Fig. 2. K_0 is the initial stiffness of cohesive element without damage ($\delta < \delta_0$). When $\delta = \delta_0$, the cohesive element damage occurs, and the traction T reaches to the peak value equal to the material tensile/shear strength. When $\delta > \delta_0$, the cohesive element experiences damage evolution, and D is the damage factor. When δ increases to δ_f , the cohesive element fails completely, forming a visible fracture. Therefore, δ_f is the fracture displacement and the area under the solid red line is equal to the fracture energy G_c .

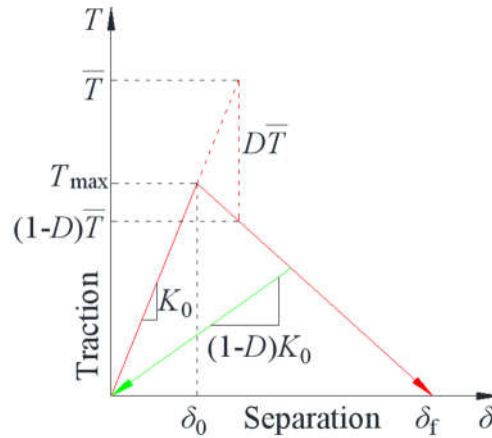


Figure 2. Irreversible bilinear traction-separation law.

The maximum nominal stress criterion is adopted to define the damage initiation in this work. This criterion assumes that the damage is to initiate when the maximum nominal stress ratio (as defined in the expression below) reaches 1.0 and can be written as

$$\max\left\{\frac{\langle T_n \rangle}{T_n^0}, \frac{T_s}{T_s^0}\right\} = 1 \quad (3)$$

Where T_n^0 and T_s^0 represent the peak values of the nominal stress when the deformation is either purely normal to the interface or purely in the shear direction, respectively.

Two types of simplified criteria based on the fracture displacement δ_f and fracture energy G_c are used to determine the tensile failure of cohesive element, and the linear friction law written as in Equation (4) is used to determine the shear failure of the cohesive element.

$$|\tau_s| = \begin{cases} f\sigma_n & f\sigma_n \leq \tau_{\max} \\ \tau_{\max} & f\sigma_n > \tau_{\max} \end{cases} \quad (4)$$

Where f is the coefficient of friction, σ_n is the normal compressive stress, τ_s is the frictional shear stress, and τ_{\max} is shear stress limit on the contacting surfaces.

2.3. Equations governing the fluid flow

The fluid constitutive response within the gap between the cohesive surfaces comprises the tangential flow and the normal flow (i.e. leakage flow), as shown in Fig. 3. The fracturing fluid is assumed to be incompressible Newtonian fluid, and its flow can be characterized using the Reynolds lubrication equation as [25, 27-28]

$$\frac{\partial w}{\partial t} + c_t(p_i - p_t) + c_b(p_i - p_b) = \frac{1}{12\mu} \nabla \cdot (w^3 \nabla p_i) + Q(t)\delta(x) \quad (5)$$

where $w(x, t)$ is the crack width, t is the injection time, x is the distance to the wellbore, c_t and c_b are fluid leak-off coefficients for the top and bottom surfaces of cohesive element, respectively. p_i is the fluid pressure, and p_t and p_b are the pore pressures in the adjacent pore fluid material on the top and bottom surfaces of the cohesive element, respectively. μ is the fluid viscosity, $Q(t)$ is the injection rate, and $\delta(x)$ is the Dirac delta function.

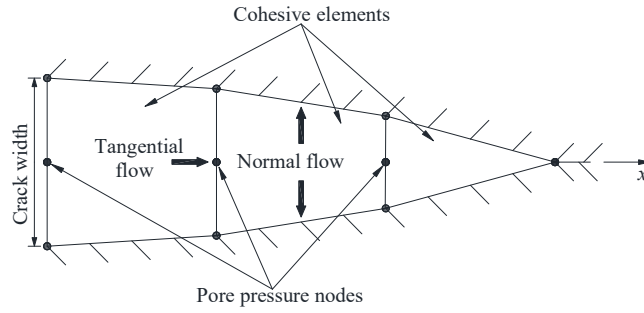


Figure 3. Fluid flow within cohesive elements.

3. Model verification

3.1. Comparison with Blanton's criteria

Aimed at the interaction between the hydraulic fracture and the pre-existing fracture shown in Fig. 4, Blanton [29-30] proposed the criteria for the opening of pre-existing fracture and the arrest of hydraulic fracture, respectively.

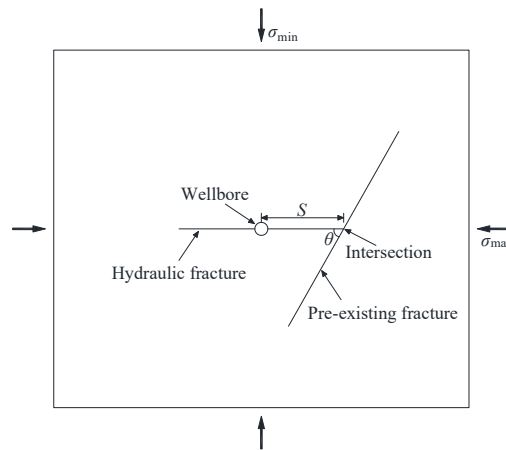


Figure 4. Diagram of hydraulic fracture intersecting pre-existing fracture.

The opening criterion can be written as

$$\sigma_{\max} - \sigma_{\min} = \left[\frac{\pi E G_c}{4(1-\nu^2)S} \right]^{1/2} \cdot \frac{1}{\sin^2 \theta} \quad (6)$$

and the arrest criterion can be written as

$$\sigma_{\max} - \sigma_{\min} = \left[\frac{\pi E G_c}{4(1-\nu^2)S} \right]^{1/2} \cdot \frac{2f}{2f \cdot \sin^2 \theta - \sin 2\theta} \quad (7)$$

Where S is the horizontal distance between the wellbore and the pre-existing fracture, and θ is the intersection angle.

The numerical model setup is shown in Fig. 5 and the major input parameters are listed in Table 1. In the series of simulations, the horizontal distance between the injection node and the pre-existing fracture is fixed as 7.5 m, the length of the pre-existing fracture is fixed as 6 m, and the initial length of the hydraulic fracture is fixed as 7 m.

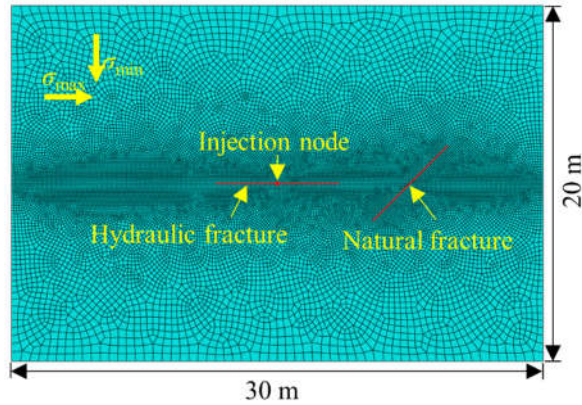


Figure 5. Model setup for interaction between hydraulic fracture and natural fracture.

Table 1. Model parameters for interaction between hydraulic fracture and natural fracture.

Input parameters	Value	Unit
Young's modulus	15	GPa
Poisson's ratio	0.25	Nondimensionless
Fluid viscosity	1	cP
Tensile strength of rock	6	MPa
Shear strength of rock	20	MPa
Tensile strength of pre-existing fracture	2	MPa
Shear strength of pre-existing fracture	10	MPa
Formation permeability	10	mD
Injection rate	0.001	m ³ /s
Specific weight of fluid	9800	N/m ³
Initial pore pressure	0	MPa
Tensile critical fracture energy for rock	150	J/m ²
Tensile critical fracture energy for natural fracture	25	J/m ²
Leak-off coefficient	1e-14	m/(Pa·s)
Porosity	0.1	Nondimensionless
Friction coefficient	0.6	Nondimensionless

The three typical interaction model, i.e. the natural fracture opening, the hydraulic fracture arrest, and the hydraulic fracture crossing, can be simulated successfully using the pore pressure cohesive element, as shown in Fig. 6. The comparison between the numerical results and the Blanton's criteria is shown in Fig. 7, from which it can be found that the numerical results are in good agreement with the analytic solution. Therefore, the pore pressure cohesive element can accurately simulate the interaction between the hydraulic fracture and the natural fracture.

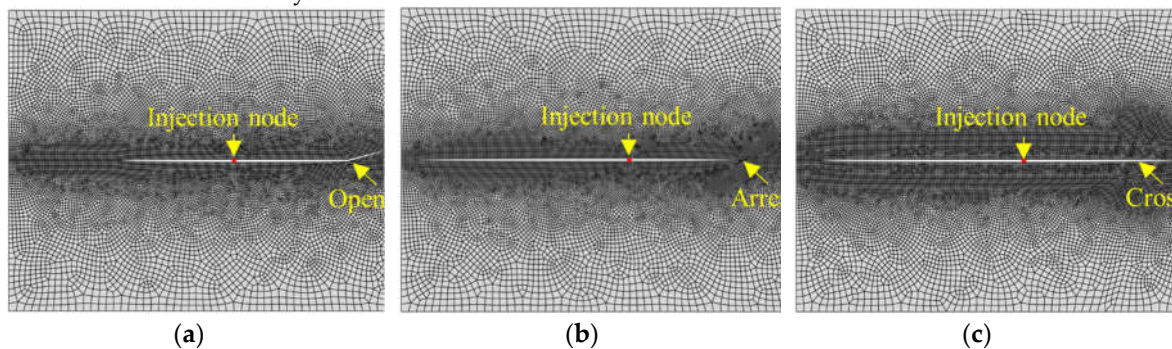


Figure 6. Interaction between hydraulic fracture and natural fracture from the present model; (a) $\sigma_{\max} - \sigma_{\min} = 0\text{MPa}$, $\theta = 15^\circ$; (b) $\sigma_{\max} - \sigma_{\min} = 2\text{MPa}$, $\theta = 45^\circ$; (c) $\sigma_{\max} - \sigma_{\min} = 4\text{MPa}$, $\theta = 75^\circ$.

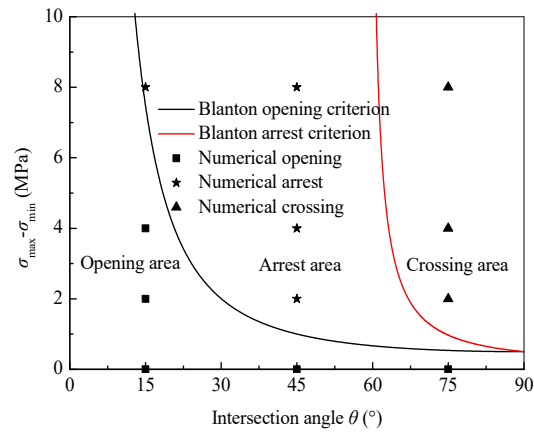


Figure 7. Comparison between numerical results and Blanton's criteria.

3.2. Comparison with indirect fracturing experiment

Using the coal-cement blocks with interfaces lubricated by oil grease or Vaseline, Jiang et al. [11] studied the effect of the stress difference and the interfacial friction on the hydraulic fracture propagation across the coal-rock interface. The physical and mechanics parameters of coal and cement blocks are listed in Table 2. According to Jiang's experiment, the numerical model is set as Fig. 8 and the input parameters are listed in Table 3. Note that, in our simulation, the shear strength of the coal-cement interface was input based on the linear friction law, and elastic modulus of the interface was input equating to half of the sum of that of the coal and the cement.

Table 2. Physical and mechanics parameters of Jiang's laboratory experiment [11].

Parameters	Coal	Cement
Porosity	0.089	0.079
Permeability (10^{-15} m ²)	0.014	0.0039
Elastic modulus (GPa)	3.48	6.58
Poisson ratio	0.23	0.19
Tensile strength (MPa)	1.69	4.56
Mode-I fracture toughness (MPa·m ^{0.5})	0.20	0.98

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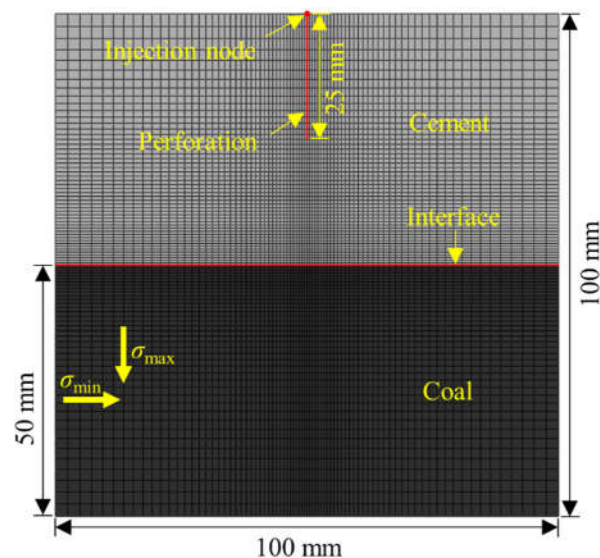


Figure 8. Model setup for indirect fracturing from cement to coal.

Table 3. Model parameters for indirect fracturing from cement to coal.

Input parameters	Coal	Cement	Interface
Porosity	0.089	0.079	—
Permeability (mD)	0.014	0.0039	—
Elastic modulus (GPa)	3.48	6.58	5.03
Poisson's ratio	0.23	0.19	—
Tensile strength (MPa)	1.69	4.56	0
Shear strength (MPa)	20	30	linear friction law
Tensile critical fracture energy (J/m ²)	10.89	140.69	0
Leak-off coefficient (m/(Pa·s))	1e-13	1e-14	Top 1e-14 Bottom 1e-13
Specific weight of fluid (N/m ³)		9800	
Initial pore pressure (MPa)		0	
Injection rate (m ³ /s)		3.33e-6	
Fluid viscosity (cP)		1	

Bulleted lists

The comparison between the numerical results and the experiment results is listed in Table 4. Note that the boldly highlighted cells show the cases at which the numerical results do not agree with the laboratory work. It can be found that the numerical results agree well with the experiment results indicating the applicability of the pore pressure cohesive element to the simulation of the hydraulic fracture propagation across the coal-rock interface.

Table 4. Comparison between numerical simulation and Jiang's experiment.

σ_{mid} (MPa)	σ_{min} (MPa)	σ_{max} (MPa)	Interfacial friction coefficient	Interfacial shear strength (MPa)	Experiment results	Numerical results
5	3	6		4.32	No crossing	No crossing
5	3	7		5.04	No crossing	No crossing
5	3	8	0.7200	5.76	No crossing	Crossing
5	3	9		6.48	Crossing	Crossing
5	3	6		2.99	No crossing	No crossing
5	3	7		3.48	No crossing	No crossing
5	3	8	0.4976	3.98	No crossing	No crossing
5	3	11		5.47	Crossing	Crossing
5	3	11		2.81	No crossing	No crossing
5	3	12		3.07	No crossing	No crossing
5	3	13	0.2557	3.32	No crossing	Crossing
5	3	14		3.58	No crossing	Crossing
5	3	15		3.84	Crossing	Crossing

4. Engineering application

4.1. Numerical model for CBM extraction using indirect fracturing

The site measurement results shows that the Xinjing mine of Shanxi province in China belongs to Strike-slip faulting stress regime, i.e. $\sigma_H > \sigma_v > \sigma_h$. The mean in-situ stress is $\sigma_v=8.74\text{MPa}$, $\sigma_H=11.34\text{MPa}$, $\sigma_h=3.55\text{MPa}$. The samples including coal, roof and floor are obtained from the 3# coal seam, the coal was cut using wire cutting machine to analyze the natural fracture distribution, as shown in Fig. 9, and the statistical result of the natural fractures in coal is listed in Table 5.



Figure 9. Natural fractures distribution in coal.

Table 5. Statistical result of the natural fractures in coal.

Geometric parameters	value	Unit
Angle between natural fracture and σ_{\min} direction	60, 120	$^{\circ}$
Length of natural fracture	0.5~2	m
Interval of natural fractures along the length direction	0.1~1	m
Interval of natural fractures vertical the length direction	0.1~1	m

Combined with the geological condition of the 3# coal, the numerical model for indirect fracturing from the roof to the coal is established, as shown in Fig. 10. Note that the natural fractures are embedded into the numerical model using python language based on the statistical result shown in Table 5. The perforation depth is set as 0.6m, and the propagation law of hydraulic fracture across the coal-rock interface is studied by changing the distance between the horizontal well and the coal, i.e. H. To minimize the dependence of fracture propagation path on the embedded location of pore pressure cohesive element and reflect truly the fracture propagation process, the pore pressure cohesive elements were globally embedded between all the solid elements.

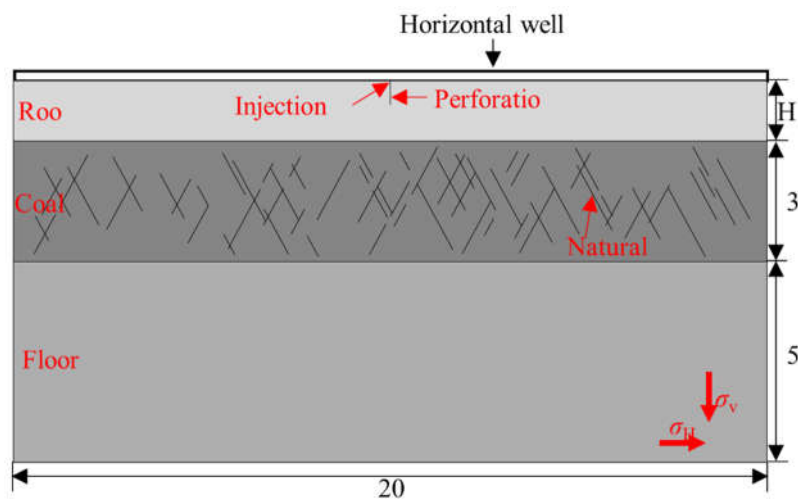


Figure 10. Model setup for CBM extraction using indirect fracturing.

The mechanical parameters of coal, roof and floor were obtained using the uniaxial compression test, Brazilian split test, and semi-circular specimens under three-point bending test. In addition, the height of hydraulic fracture was estimated as 5m and the injection rate in the simulation was set as $0.001\text{m}^3/\text{s}$. Considering the actual fracturing parameters with perforation interval of 3m and perforation density of 10 in 1m, the injection rate of $0.001\text{m}^3/\text{s}$ in simulation is correspond to that of

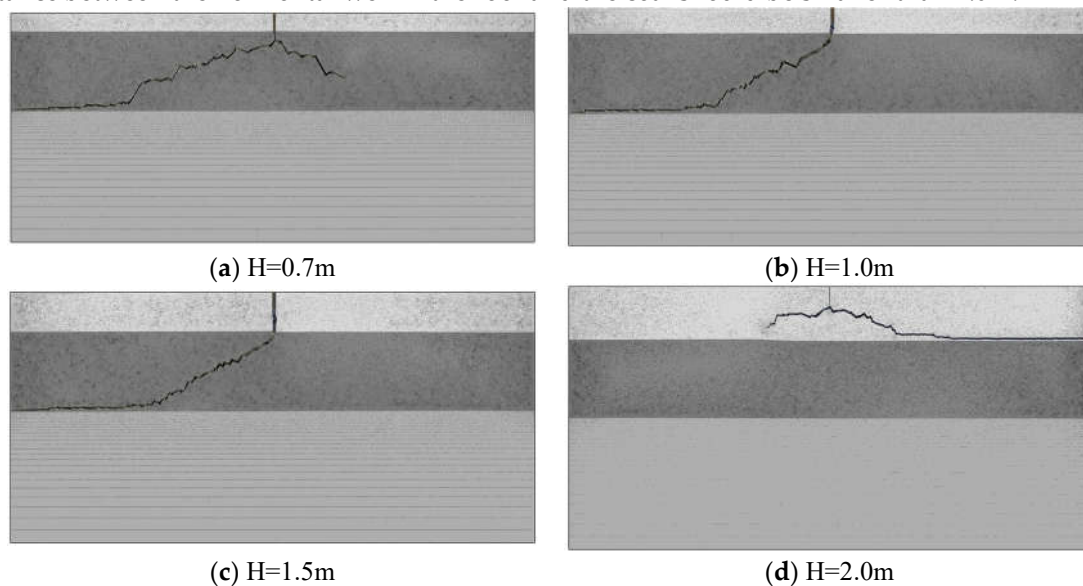
9m³/s in actual engineering. The specific input parameters for the numerical model are listed in Table 6.

Table 6. Model parameters for CBM extraction using indirect fracturing.

Input parameters	Roof	Coal	Floor
Elastic modulus (GPa)	5.65	1.35	19.57
Poisson's ratio	0.25	0.3	0.2
Permeability (mD)	0.1	1	1
Porosity	0.1	0.2	0.1
Tensile strength (MPa)	3.51	0.6	4.44
Shear strength (MPa)	25	20	30
Tensile strength of natural fracture (MPa)	—	0.001	—
Shear strength of natural fracture (MPa)	—	10	—
Fracture displacement (m)	0.001	0.0015	0.0005
Fracture displacement of natural fracture (m)	—	0.00075	—
Leak-off coefficient (m/(Pa·s))	1e-14	1e-13	1e-14
Specific weight of fluid (N/m ³)		9800	
Fluid viscosity (cP)		1	
Injection rate (m ³ /s)		0.001	
Initial pore pressure (MPa)		5.62	

4.2. Results analysis

The propagation paths of hydraulic fracture for different values of H are shown in Fig. 11, from which it can be found that, when H=0.7m, 1.0m, and 1.5m, the hydraulic fracture can propagate into the coal from the roof, when H≥0.7m, the hydraulic fracture can not propagate into the coal. Therefore, in terms of the indirect fracturing for CBM extraction in 3# coal of Xinjing mine, the distance between the horizontal well in the roof and the coal should be smaller than 2.0m.



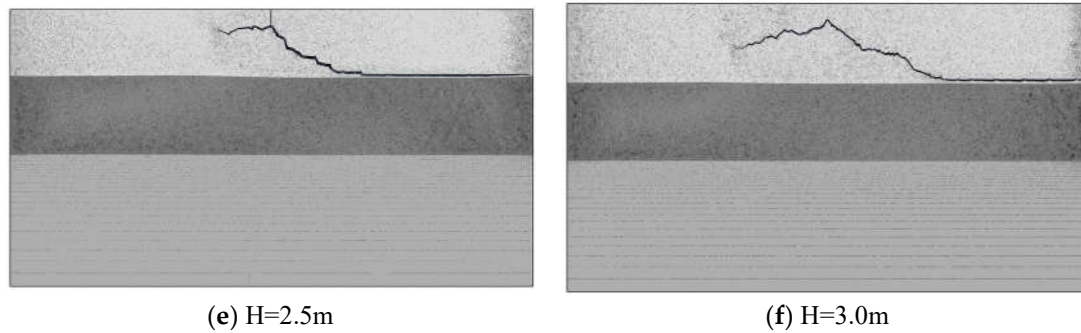


Figure 11. Propagation path of hydraulic fracture under different horizontal well drilling position.

The injection pressure-injection time curve under different horizontal well drilling position was extracted, as shown in Fig. 12, from which the perforation breakdown pressure and the crossing pressure (i.e. the injection pressure at the time when the hydraulic fracture just propagates to the interface) were recorded in Table 7, respectively.

It can be found from Fig. 11 and Table 7 that the perforation breakdown pressure increases with increasing H and this phenomenon is more obvious when H is small ($H=0.7\text{m}$, 1.0m and 1.5m). The mechanical mechanism for this phenomenon is that the incompatible deformation between roof and coal induces the tensile internal stress when fracturing, and the induced tensile stress increases with decreasing H . Therefore, as the horizontal well drilling position H increases, the breakdown pressure increases but increment decreases.

It can be found from Tabel 7 that, for $H=0.7\text{m}$, 1.0m and 1.5m , the hydraulic fracture can propagate into the coal but the crossing pressure decreases with increasing H , the reason for this is that when H is large, the released elastic strain energy due to the roof fracture is more, the part of the elastic strain energy is transformed into the surface energy of hydraulic fracture, and the other is consumed to overcome the fracture propagation resistance, keeping the continuous propagation of hydraulic fracture in coal.

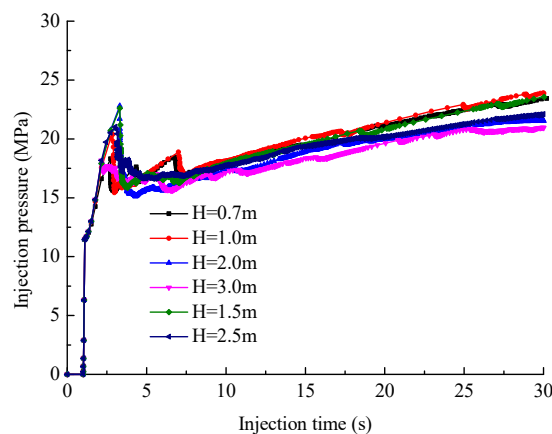


Figure 12. Injection pressure under different horizontal well drilling position.

Table 7. Fracturing characteristic parameters under different horizontal well drilling position.

Horizontal well drilling position H (m)	Perforation breakdown pressure (MPa)	Crossing pressure (MPa)
0.7	16.60	17.55
1.0	17.55	16.54
1.5	17.91	16.06
2.0	18.02	—
2.5	18.17	—
3.0	18.18	—

5. Conclusion

1. Pore pressure cohesive element can be used to accurately simulate the propagation of the hydraulic fracture across the coal-rock interface in coal measure strata and the interaction between hydraulic fracture and natural fracture.
2. As the horizontal well drilling position decreases, the perforation breakdown pressure decreases while the crossing pressure increases. The mechanical mechanism for this phenomenon is that the incompatible deformation between roof and coal induces the tensile internal stress when fracturing, and the induced tensile stress increases with decreasing H.
3. Considering the co-effect of natural fractures and tectonic stress in 3# coal of Xinjing mine of Shanxi province in China, it is suggested that the horizontal well drilling position is smaller than 2.0m when indirectly fracturing the coal from roof, the hydraulic fracture can propagate into the coal successfully.

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