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Experimental Investigation of Stress Rate and Grain Size on Gas Seepage Characteristics of Granular Coal

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Abstract: Coal seam gas, held within the inner pores of unmineable coal, is an important energy resource. Gas release largely depends on the gas seepage characteristics and their evolution within granular coal. To monitor this evolution, a series of experiments were conducted to study the effects of applied compressive stress and original grain size distribution (GSD) on the variations in the gas seepage characteristics of granular coal samples. Grain crushing under higher stress rates was observed to be more intense. Isolated fractures in the larger diameter fractions transformed from self-extending to inter-connecting pathways at a critical compressive stress. Grain crushing was mainly caused by compression and high-speed impact. Based on the test results of the original GSD effect, the overall process of porosity and permeability evolution during compression can be divided into three different phases: (1) rapid reduction in the void ratio; (2) continued reduction in the void ratio and large particle crushing; and (3) continued crushing of large particles. Void size reduction and particle crushing were mainly attributed to the porosity and permeability decreases that occurred. The performance of an empirical model, for porosity and permeability evolution, was also investigated. The predictive results indicate that grain crushing caused permeability increases during compression, and that this appeared to be the main cause for the predictive values being lower than those obtained from the experimental tests. The predictive accuracy would be the same for samples under different stress rates and the lowest for the sample with the highest proportion of large grain diameters.

Keywords: gas permeability; granular coal; porosity; stress rate; grain size distribution; grain crushing

1. Introduction

Coal seam gas, held within the inner pores in unmineable coal, is an important energy resource. However, as shown in Figure 1, longwall mining may commonly form four zones of disturbance after disturbing the overlying strata of a coal seam [1]; this makes the coal seam gas within the coal be rapidly released into the gob. Gas releases from coal during mining not only causes engineering accidents (gas outbursts, coal and gas outburst, etc.) [2,3], but also contributes to global warming by the greenhouse effect [4]. The permeability increased by up to three orders of magnitude in the fractured (caved) zone and is usually several orders of magnitude greater than that of the intact coal [5]. Such a high permeability in coal threatens the normal storage and transport of gas within the coal seam [6].

Significant experimental, numerical, or theoretical methods were conducted to understand the properties of granular coal, such as density, ash, and particle size effects on the fixed characteristics [7–11], coal particle moving [12], fragmentation behavior [13], coal combustion [14], etc. Adánez et al. [7] conducted a series of experiments and proposed an equation to evaluate the transport velocities of sand and coal particles. For coal blocks having larger particle diameters, the gas desorbs from its micropores, diffuses into macropores, and then seeps under a pressure gradient [15,16]. Hu et al. [17] experimentally and numerically investigated the scale effects and formation mechanism of gas releases from coal particles, and a bidisperse diffusion model was established to predict the experimental results; they found that the scale effects of gas releases from coal are controlled by the multi-scale pore structure of coal. In studies conducted using coal core samples as opposed to granular coal, it was found that under constant total stress conditions, the permeability for adsorbed gases increases when the pore pressure is reduced due to coal swelling [18–20], and decreases with increasing pore pressure due to matrix shrinkage [21].

Gas permeability is also influenced by fracture geometry [6,22], fracture geometry and water-content [23], and the presence of water [24]. The permeability evolution of granular coal was tested and the particle size [25] and temperature [26] were proven as affecting factors. According to the experimental studies of different rocks [27,28], altered permeability strongly relates to the original particle size distribution. It is clear that the seepage characteristics of granular coal are highly affected by the stress condition and granular size distribution. In order to understand the impact on seepage characteristics evolution, we first present the test coal samples and experimental equipment in Section 2. The details of the testing procedure and measurement principle for porosity and permeability evolution are given in Section 3. Section 4 analyses the test data for the compressive stress state and grain size mixture effect on the seepage characteristics and particle crushing during compression, and builds a model for porosity and permeability prediction.

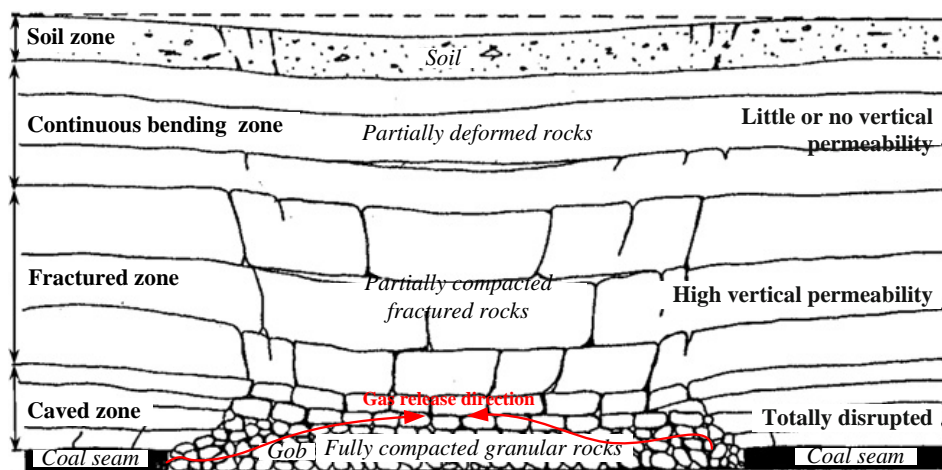


Figure 1. Overburden strata and granular coal distribution above a longwall panel [1].

2. Granular Coal Samples and Testing Equipment

2.1. Preparation of Granular Coal

Lightly weathered coal blocks were taken from the coal seam No. 2 of the Yuheng coal field in Yulin City of China. The dry density of the coal blocks was $\rho_c = 1.65 \times 10^3 \text{ kg/m}^3$. The coal blocks were first hammered into particles of less than 20 mm diameter. The granular coal particles were then separated into seven size ranges: 20–15 mm, 15–12 mm, 12–10 mm, 10–8 mm, 8–5 mm, 5–2.5 mm, and <0.25 mm. These groups were mixed according to the well-known Fuller equation in Equation (1) [29] to produce several different particle gradings.

$$p_j = (d_j/D)^n \times 100\% \quad (1)$$

where j is the size range of the granular coal, and its maximum value is 7 according to the sample preparation; D is the largest grain diameter, and $D = 20$ in this test; d_j is the largest grain diameter in the range of j sample; p_j is the mass ratio of the size range of the j sample; and n is the control parameter of the grain size distribution (GSD) for the sample, and then the seven granular ranges are mixed by $n = 0.4, 0.7, 1, 1.3,$ and 1.6 , respectively. The GSD curve for the five mixtures is shown in Figure 2.

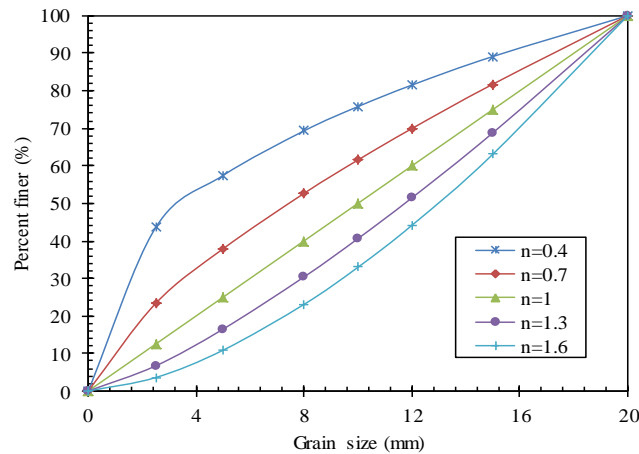


Figure 2. Grain size distribution (GSD) curves of tested granular coal samples.

2.2. Sample Arrangement

The gas adsorbate used was N_2 , with a bulk gas density of $\rho_g = 1.16 \text{ kg/m}^3$ and kinetic viscosity of $\mu_g = 1.76 \times 10^{-5} \text{ Pa}\cdot\text{s}$ under laboratory conditions ($25 \text{ }^\circ\text{C}$ at 100 kPa). As shown in Figure 3 and Table 1, in Test 1, samples (a, b, and c) were used to monitor the effect of the compressive stress rate on the seepage characteristics; their size mixtures were the same, i.e., $n = 1$; the stress rate for paths OA, OB, and OC were 0.01 kN/s (sample a), 0.02 kN/s (sample b), and 0.04 kN/s (sample c), respectively, and the corresponding loading times were t_1 , t_2 and t_3 . According to the in-situ conditions of the Yuheng coal field, the buried depth z of the coal seam is about 320 m , based on $\sigma = \gamma z = \rho_s g z$, where $\rho_s \approx 2500 \text{ kg/m}^3$ is the density of overburden strata. Therefore, the compressive stress σ was set as 8 MPa , where $t_1 = 2t_2 = 4t_3$. In Test 2, samples (d, e, f, g, and h) were used to monitor the original GSD effect; their loading paths were the same, i.e., the stress rate was 0.02 kN/s ; the original GSD for samples d, e, f, g, and h were $n = 0.4, 0.7, 1, 1.3,$ and 1.6 , respectively. The compressive stress of 8 MPa was achieved at t_2 .

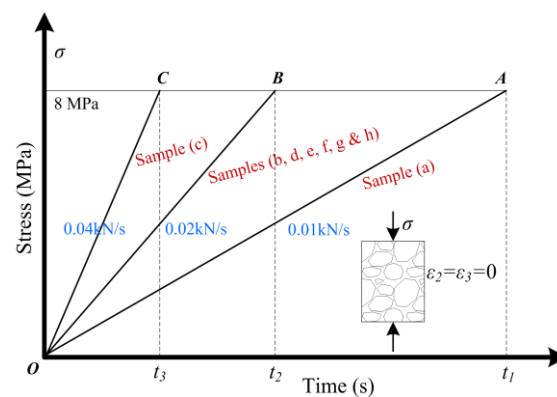


Figure 3. Loading paths.

Table 1. Stress rate and mass distribution of the particle size arrangement of each sample.

Test No.	Sample No.	Stress Rate (kN/s)	Particle Mixtures (n in Equation (1))	Weight to Each Particle Size (g)						
				20–15 mm	15–12 mm	12–10 mm	10–8 mm	8–5 mm	5–2.5 mm	2.5–0 mm
1	a	0.01	1	300	180	120	120	180	150	150
	b	0.02								
	c	0.04								
2	d	0.02	0.4	130.4	91.3	68.8	77.7	142.6	166.9	522.3
	e		0.7	218.9	141.9	100.6	106.8	177.1	174.8	279.9
	f		1	300	180	120	120	180	150	150
	g		1.3	374.4	207.9	130.3	122.7	166.7	117.5	80.4
	h		1.6	442.7	227.4	134.1	118.9	146.4	87.5	43.1

2.3. Testing System Design

The in-house testing system principle of operation is illustrated in Figure 4. The gas seepage testing box produces only a 5% pressure drop for a 5 MPa gas per 10 min. MTS815.02/286.31 [30] was used to provide the compressive stress for controlling the height of the coal sample (G4). In the gas seepage apparatus, a loading plate (G1) was utilized to maintain the compressive stress. Porous disks (G6) and (G9) enable the gas to seep freely. An epoxy resin separation layer (G11) separates the coal specimen (G4) and the vessel (G12), and prevents annular gas seepage. The triaxial base of the system and cylinder tube are connected by a one-way valve, which includes the base plate (G7), valve chest (G8), mechanical bolt (G14), mechanical spring (G15), and valve core (G16). O-rings (G2) and (G13) were utilized to enclose the gaps between the piston (G3) and the vessel (G12), base plate (G7), and valve chest (G8).

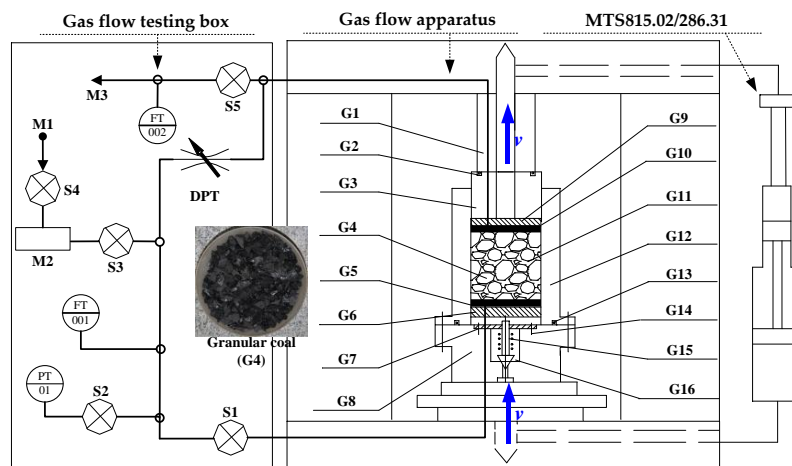


Figure 4. Testing principle of gas permeability of the granular coal. Note: G1—loading plate; G2,G13—O-rings; G3—piston; G4—granular coal sample; G5,G10—felt pad; G6,G9—porous disk; G7—base plate; G8—valve chest; G11—epoxy resin separation layer; G12—vessel; G14—mechanical bolt; G15—mechanical spring; G16—valve core; M1—gas in; M2—accumulator; M3—gas out; and S1–S5—switch.

3. Measurement Principle and Testing Procedure

3.1. Porosity Measurement

As shown in Figure 5, based on the vessel height H_a (180 mm), the deviator height H_b (110 mm), the porous disk thickness H_c (9 mm), the felt pad thickness H_d (2 mm), H_{ei} was the height of the deviator that exceeds the vessel at compressive stress σ_i .

The value H_{ei} at compressive stress σ_i can be measured by linear variable displacement transformer (LVDT) during loading. Therefore, the porosity ϕ_i at each sample height H_i at compressive stress σ_i can be calculated by Equation (2):

$$\phi_i = 1 - \frac{m}{\pi r^2 \rho_c H_i} = 1 - \frac{m}{\pi r^2 \rho_c (H_{ei} + 48)} \quad (2)$$

where m and ρ_c are the mass and density of the granular coal sample, respectively, r is the radius of the cylindrical tube, which in the test was $r = 50$ mm. The porosity variation at compressive stress σ_i was calculated by Equation (3):

$$\Delta\phi_i = \phi_{i-1} - \phi_i \quad (3)$$

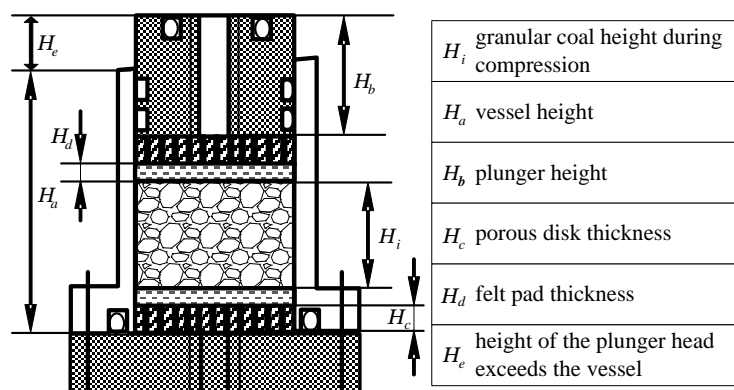


Figure 5. Sketch calculating the height of the granular coal.

3.2. Permeability Measurement

For Darcy seepage, Equation (4) describes the relation between the pore pressure gradient $\partial p / \partial h$ and the gas seepage velocity v :

$$-\partial p / \partial h = \mu_g \zeta^{-1} v \quad (4)$$

where p is the pore pressure, h is the vertical axis going through the center of the sample, μ_g is the gas kinetic viscosity, and ζ is the permeability of the granular coal. The Darcy equation (Equation (4)) can be used to model the gas flow in the granular coal sample, which is further verified by the Reynolds number (Re) [31]:

$$Re = \frac{\rho_g v d \phi_i}{\mu_g} \quad (5)$$

where d is the characteristic size, and for the non-consolidated granular coal sample, d is the average diameter of the grains, while for the consolidated granular sample, d is the average value of the capillary diameters. Darcy flow follows if Re is lower than 10 [32].

In this test, the gas flux and pore pressure are all measured for the calculation of the seepage characteristics. When gas compressibility was considered, we assume p_{in} , ρ_{in} , and v_{in} were the pore pressure, gas density, and velocity at the intake boundary (gas in), respectively. The downstream end was connected to the outtake boundary p_{out} , ρ_{out} , and v_{out} (gas out), respectively.

The gas equation of state is:

$$p = \rho_g R T \quad (6)$$

where, R is the universal gas constant and T is the temperature of the sample.

The continuity equation is:

$$Q_g = \rho_g A v \quad (7)$$

where Q_g is the gas mass flow rate, ρ_g is the gas density, A is the cross-sectional area of the sample, and $A = \pi r^2$ in this test.

For simultaneous Equations (6) and (7), we have:

$$v = \frac{Q_g RT}{A p} \quad (8)$$

Substituting Equation (8) into Equation (4), we have:

$$-\frac{\partial p}{\partial h} = \frac{\mu_g}{\zeta} \frac{Q_g RT}{A} \frac{1}{p} \quad (9)$$

Let $c_1 = \frac{\mu_g}{\zeta} \frac{Q_g RT}{A} = pv \frac{\mu_g}{\zeta}$, then integrate Equation (9), to obtain:

$$\frac{1}{2} p^2 = -c_1 h + c_2 \quad (10)$$

Substitute the boundary condition $p = p_{in}$ when $h = 0$ into Equation (10), and we obtain $c_2 = \frac{1}{2} p_{in}^2$. Then,

$$p^2 = p_{in}^2 - 2pv\mu_g\zeta^{-1}h \quad (11)$$

Therefore, the permeability ζ_i at each sample height H_i at compressive stress σ_i can be calculated by:

$$\zeta_i = \frac{2p_{i-in}v_{i-in}\mu_g H_i}{p_{i-in}^2 - p_{i-out}^2} \quad (12)$$

where p_{i-in} and v_{i-in} are the pore pressure and gas velocity at the intake boundary (gas in), and p_{i-out} is the pore pressure at the outtake boundary (gas out) at the compressive stress σ_i . That means the permeability ζ_i can be calculated by testing p_{i-in} , v_{i-in} , and p_{i-out} at the compressive stress, σ_i .

3.3. Testing Procedure

The experimental testing strictly observed the following procedural steps:

- (1) Regulate the initial height (porosity) of granular coal as a designed value before loading.

At the initial stage, the specimen height was set at 110 mm, which means the initial height $H_0 = 48 + H_{e0} = 130$ mm. Therefore, the initial height H_{e0} was fixed as 82 mm. Based on Equation (2), the initial porosity of each sample was $\phi_0 = 0.288$.

- (2) Apply and maintain the compressive stress and gas pressure to the sample.

As shown in Figure 3, in test 1, we applied the stress rate in paths OA, OB, and OC as 0.01 kN/s (sample a), 0.02 kN/s (sample b), and 0.04 kN/s (sample c), respectively, and the gas in pressure at the intake boundary was fixed at 0.600 MPa; in test 2, we applied the stress rate in path OB as 0.02 kN/s (samples d, e, f, g, and h), the gas in pressure at the intake boundary was also fixed as 0.600 MPa.

Meanwhile, the compressive stress and gas pressure were applied to the coal sample and kept steady until the completion of step 3. In this step, the displacement of the coal sample could be measured by the LVDT sensors. The porosity and its variation at each time interval could be calculated by Equations (2) and (3). Meanwhile, $p_{i-in} = p_{in}$ was fixed at 0.600 MPa, and the v_{i-in} and p_{i-out} measurement values were used to calculate the permeability based on Equation (12).

- (3) The test was ended when the compressive stress met the designed value.

The final compressive stress for each test was fixed as 8 MPa, based on Equation (13):

$$d\sigma/dt = \frac{dF/dt}{\pi r^2} \quad (13)$$

where, $r = 50$ mm, and dF/dt for the samples (a, b and c) were 0.01 kN/s, 0.02 kN/s, and 0.04 kN/s respectively, therefore the time to close the test was $t_1 = 2t_2 = 4t_3 = 6280$ s.

The entire test procedure is shown in Figure 6. To check the repeatability and maintain accuracy during the test, each coal sample was recorded as the mean average of three repeat tests.

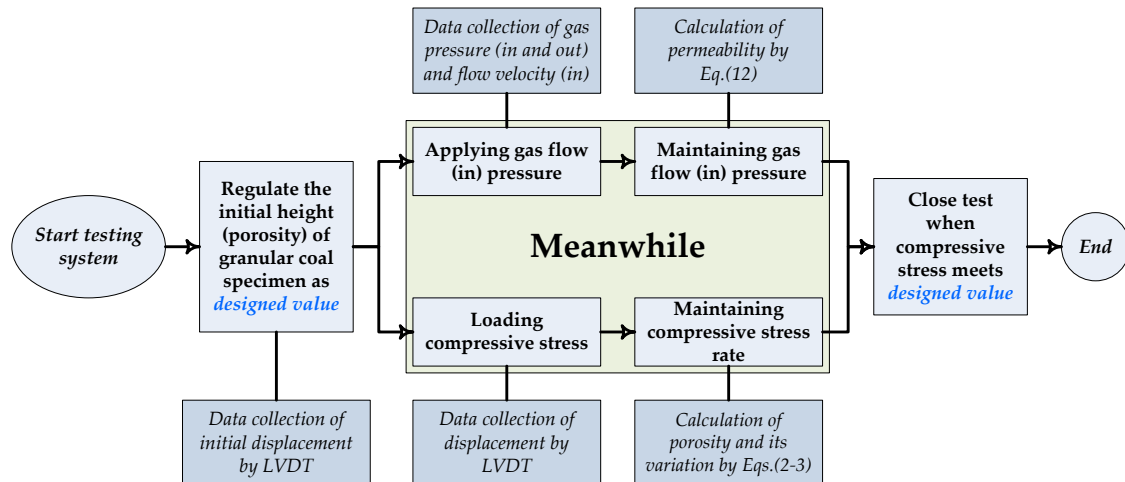


Figure 6. Testing procedure.

4. Testing Results and Discussion

In Kong's research [31], the upper limit of Re suitable for the Darcy law was 10 and for different porous media this limit is slightly different. We calculated the Re according to Equation (5) to prove that the Darcy equation (Equation (4)) can be used to model the gas flow in granular coal samples. Sample (a) was taken as an example; the average diameter d_{50} was 10 mm at the initial stage and it will decrease during crushing. Therefore, Re can be calculated based on Equation (5) by considering $d_{50} = 10$ mm. As shown in Table 2, the porosity, gas in flow rate, gas out pressure, Re , and permeability were calculated. The maximum Re for sample (a) is 8.095, which is lower than 10 and Darcy flow follows.

Table 2. Re and permeability test results of sample (a).

Compressive Stress (MPa)	Sample Height (mm)	Porosity	Gas in Flow Rate ($\text{mm}\cdot\text{s}^{-1}$)	Gas Out Pressure (MPa)	Re	Permeability (μm^2)
0	130	0.288	42.689	0.587	8.095	7.488
1	111.9	0.173	32.902	0.553	3.751	1.422
2	106.6	0.131	19.366	0.531	1.674	0.557
3	104.5	0.114	24.573	0.553	1.851	0.717
4	103.1	0.102	17.528	0.498	1.181	0.341
5	102.1	0.093	10.909	0.462	0.671	0.161
6	101.4	0.087	9.152	0.455	0.521	0.128
7	100.8	0.081	4.461	0.476	0.238	0.071
8	100.3	0.077	3.469	0.129	0.175	0.021

4.1. Effect of Stress Rate on the Seepage Characteristics Evolution

4.1.1. Effect of Compressive Stress Rate on the Porosity and Permeability Evolution

The porosity–compressive stress curve and porosity variation–compressive stress curve for samples (a, b and c) are presented in Figure 7, based on Equations (2) and (3), respectively. Figure 8 illustrates the relationship between the permeability evolutions of samples (a, b, and c). It can be seen from these figures that in general the porosity, permeability, and void space all decrease during

compression. At the same compressive stress, the porosity and permeability of coal samples under higher stress rates appear larger, which means the porosity and permeability decrease to a lesser degree. Compared with the porosity and permeability at the beginning (compressive stress: 0 MPa), the final porosity (compressive stress: 8 MPa) of samples (a, b, and c) decrease by 73.37%, 55.54%, and 44.35%, respectively. The corresponding final permeability values were $0.021 \mu\text{m}^2$, $0.751 \mu\text{m}^2$, and $1.616 \mu\text{m}^2$, respectively, where the value of sample (c) is almost two orders of magnitude greater than that of sample (a).

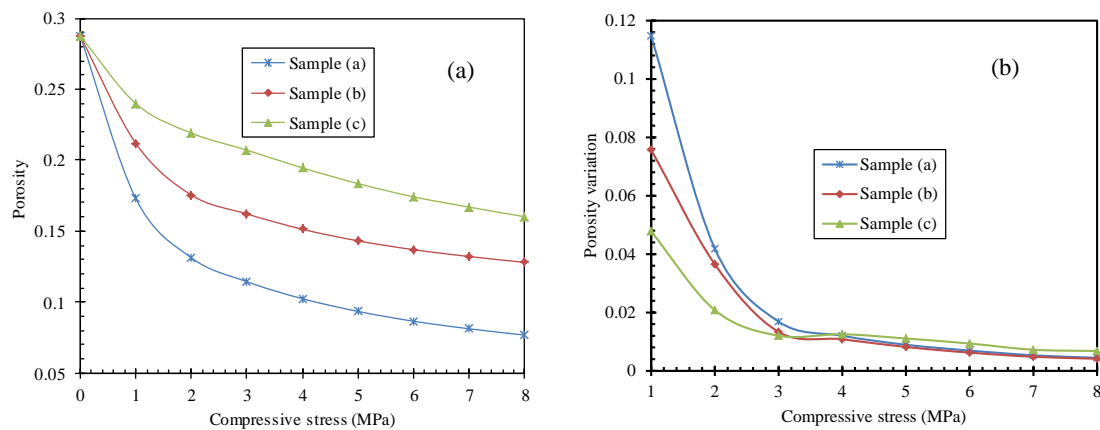


Figure 7. Porosity (a) and its variation (b) of samples (a, b, and c) during compression.

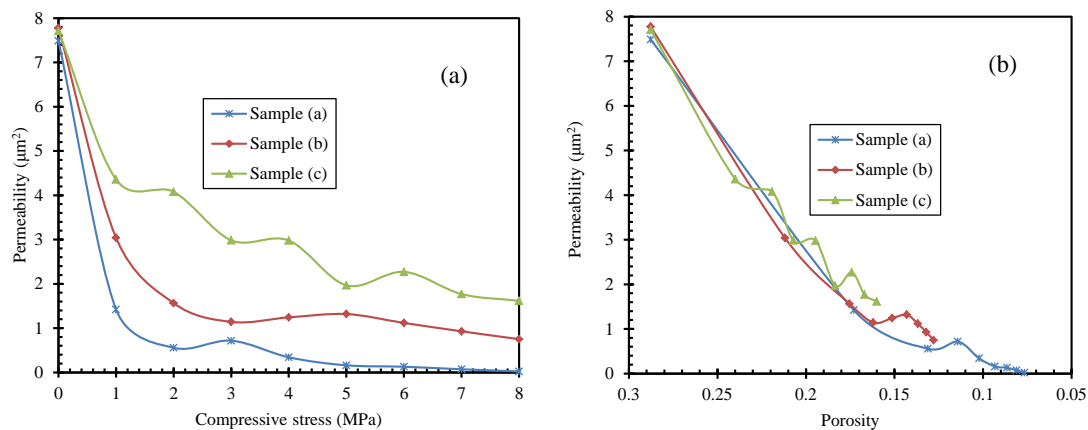


Figure 8. Permeability evolutions for samples (a, b, and c). (a) Permeability–compressive stress curves; and (b) permeability–porosity curves.

Figure 7b shows that the porosity variation decreases during compression, due to a progressive reduction in void diameter. Four MPa appeared to be the critical compressive stress for porosity variation, and the porosity variation for sample (a) was larger than that for the other samples before the critical compressive stress, which means that the voids in sample (a) were wider than those in the other samples. The porosity variation as well as the void diameter in the three samples was almost the same at the critical compressive stress.

As shown in Figure 8, the permeability under the larger stress rate has higher values due to the sample being subjected to a larger stress rate. This most likely occurs because the compressive stress causes a greater degree of closure of the void space. However, there were some abnormal increases in permeability at the critical compressive stress. As seen in Figure 8, the critical compressive stress for samples (a, b, and c) was 2–3 MPa, 4–5 MPa, and 5–6 MPa, respectively, with a corresponding porosity of 0.13–0.11, 0.15–0.14, and 0.18–0.17. At the critical compressive stress, isolated fractures in the larger

grain size sample achieved the transformation from self-extension to inter-connection, which increases the permeability rapidly.

4.1.2. Effect of Stress Rate on the Grain Size Distribution after Crushing

A re-distribution of particle sizes was observed after testing. Using sieve analysis, the GSD was determined as shown in Table 3, and the variation is shown in Figure 9, compared with the original GSD (Test 1) in Table 1. The mass of <5 mm diameter grains was increased due to attrition of the particles in the 5–20 mm range. The mass of the size fraction 20–8 mm decreased, but increased for the <5 mm size fraction, when at a lower stress rate. The fraction of particles in the 8–5 mm size range reduced for samples (a and b) but increased for sample (c); which indicates that particle crushing could be occurring by compression and fracture.

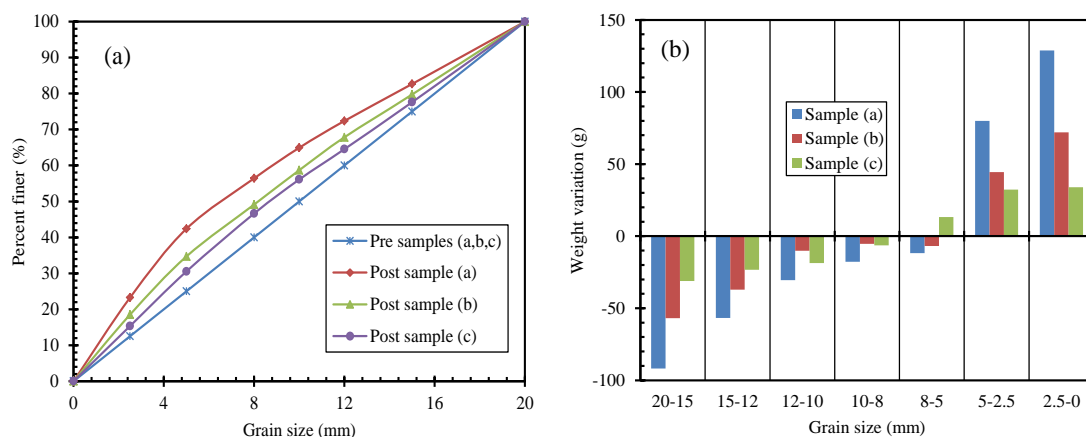


Figure 9. Grain size distributions and weight variation for samples (a, b, and c). (a) Grain size distribution; and (b) weight variation after testing (+ means increased, – means decreased).

Table 3. Mass distribution of particle size after testing under variable stress rate.

Sample No.	Stress Rate (kN/s)	Weight to Each Particle Size (g)						
		20–15 mm	15–12 mm	12–10 mm	10–8 mm	8–5 mm	5–2.5 mm	2.5–0 mm
a	0.01	208.1	123.3	89.5	102.2	168.2	229.9	278.8
b	0.02	243.2	142.8	109.8	114.6	173.2	194.4	222.0
c	0.04	268.9	156.7	101.3	113.5	193.3	182.4	184.0

4.2. Effect of Original Grain Size Distribution on the Seepage Characteristics

4.2.1. Effect of Original Grain Size Distribution on the Porosity and Permeability Evolution

The porosity–compressive stress curve and porosity variation–compressive stress curves (for variable size mixtures) are presented in Figure 10. The overall permeability–compressive stress (through porosity) curves presented in Figure 11a–c show the permeability–porosity curves and the permeability–compressive stress curves after 2 MPa. The overall process of porosity and permeability evolution during compression can be divided into three different phases: rapid reduction in the void ratio, continued reduction in the void ratio and large particle crushing, and continued crushing of the large particles.

The first phase involves the rapid reduction in the void ratio during which the porosity and permeability decrease by up to 40% (Figure 10a). When the compressive stress is less than 3 MPa during compression, the porosity of the samples with a higher proportion of larger particles (e.g., 20–10 mm diameter) appears to be larger. This suggests that their porosity and permeability decrease by a smaller amount, which is most likely due to the lower efficiency in particle packing.

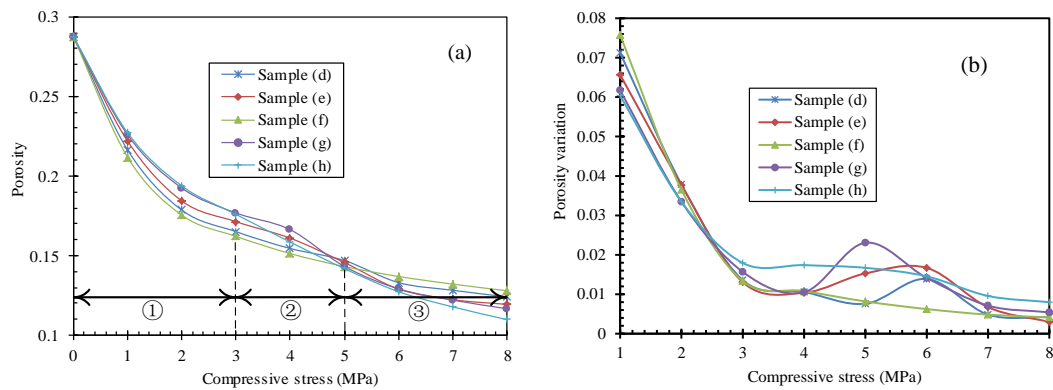


Figure 10. Porosity (a) and its variation (b) of each sample within variable particle sizes.

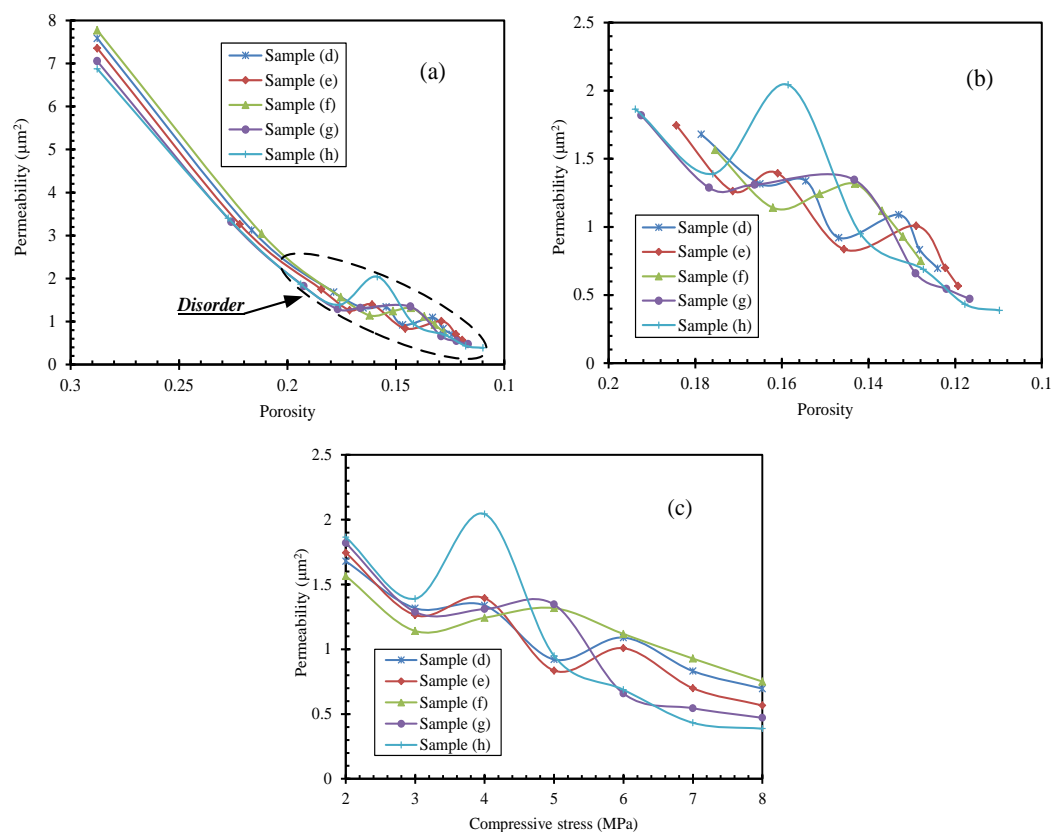


Figure 11. Permeability evolutions during compression. (a) Whole process; (b) permeability–porosity after 2 MPa; and (c) permeability–compressive stress after 2 MPa.

The second stage occurs when void shrinkage continues, and crushing of the larger particles begins during which the largest particles (e.g., 20–15 mm diameter) were crushed during compression. This occurred when the compressive stress was between 3 and 5 MPa. The porosity in sample (h) decreased rapidly (see Figure 10a) whilst its porosity variation was larger than that of the other samples (see Figure 10b). This indicates that the largest particles (e.g., 20–15 mm) in sample (h) were crushed at a faster rate.

The third stage of porosity evolution involves continuous crushing of the large particles. It can be seen from Figure 10b that the rate of change in porosity decreases during compression. The rate of porosity decrease was reduced with progressive compression due to the monotonous decrease in the void diameter. Five MPa was the critical compressive stress for alteration of both the porosity

(Figure 10b) and permeability (Figure 11c). Porosity alteration for sample (f) was at a minimum after the critical compressive stress, which means that the inter-particle pore diameter was at its smallest value. Furthermore, the bulk porosity of each sample was almost the same as the critical compressive stress (5 MPa), when compared with the porosity at the critical compressive stress. The porosity at a compressive stress of 8 MPa was lower for samples having a higher fraction of large particle diameters. The porosity and permeability of sample (f) were at their maximum; this means that the porosity decrease is mainly contributed by particle crushing in the last phase.

To summarise, the original GSD was a key factor for determining both the porosity and permeability evolution during compression. The average reduction of porosity in the first two phases was 48%, but in the third phase was only 8%. It appears that the inter-particle porosity between smaller diameter particles was more prone to volume reduction than for the larger particles. Larger particles were more prone to crushing, which led to significant decreases in porosity in the third phase of compression.

4.2.2. Effect of Original Grain Size Distribution on the Post-Crushing Grain Size Distribution

Table 4 shows the typical GSD after testing and their variation is shown in Figure 12, compared with the original GSD (Test 2) in Table 1. The mass of the <5 mm size fraction was increased after compression for each sample. However, the mass of the 20–8 mm size fraction was reduced. During compression, larger grains (e.g., 20–8 mm) were broken into smaller sizes, which caused the mass of the <5 mm size fraction to increase. Furthermore, the 20–15 mm size fraction decreased the most and this was attributed to larger particles being more easily crushed during compression. This is why the critical compressive stress for sample (h) shows a minimum value in Figure 11c. Samples with a higher proportion of larger diameter particles showed a larger decrease in the mass of the 20–12 mm size fraction and an increase in the <5 mm size fraction. The fraction of particles in the 8–5 mm size range increased for samples (d and e), but reduced for samples (f, g, and h).

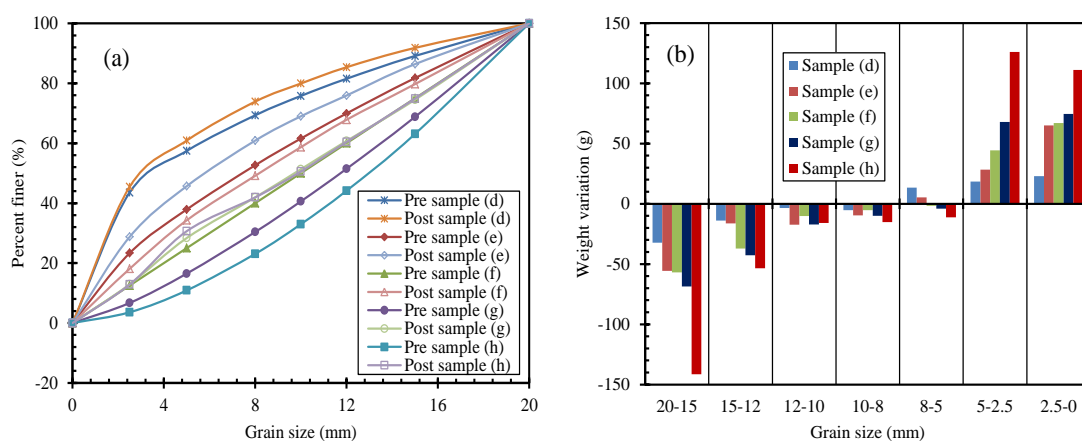


Figure 12. Grain size distributions and weight variation for samples within the variable original GSD. (a) Grain size distribution; and (b) weight variation after test (+ means increased, – means decreased).

Table 4. Mass distribution of the particle size after testing with the variable original GSD.

Sample No.	Particle Mixtures (<i>n</i> in Equation (1))	Weight to Each Particle Size (g)						
		20–15 mm	15–12 mm	12–10 mm	10–8 mm	8–5 mm	5–2.5 mm	2.5–0 mm
d	0.4	98.3	77.4	65.2	72.4	156.0	185.4	545.3
e	0.7	163.2	125.7	83.2	97.3	182.5	203.2	344.9
f	1	243.2	142.8	109.8	114.6	178.2	194.4	217.0
g	1.3	305.8	165.2	113.2	112.8	162.7	185.4	154.9
h	1.6	301.2	173.9	118.2	103.7	135.3	213.5	154.2

4.3. Porosity and Permeability Prediction during Testing

Due to the general lack of published data on the evolution of permeability and porosity for granular coals, it is common practice to predict the permeability and porosity using an empirical equation and some very limited experimental data. Chilingar [33] defined the relationship between porosity and permeability in porous media (sands and sandstones) as:

$$\zeta_i = \frac{d_e^2 \phi_i^3}{72(1 - \phi_i)^2} \quad (14)$$

where, d_e is the effective diameter of grain sizes. According to Equation (14), we obtain:

$$\frac{\zeta_i}{\zeta_0} = \left(\frac{\phi_i}{\phi_0} \right)^3 \left(\frac{1 - \phi_0}{1 - \phi_i} \right)^2 \quad (15)$$

where, ϕ_0 is the initial porosity and ζ_0 is the initial permeability. The test and predictive data of permeability were shown in Figure 13. A simple statistical efficiency criterion was used to analyze the model performance in Equation (15), based on the coefficient of determination factor R^2 :

$$R^2 = \frac{\sum_{i=1}^l (\zeta_i^t)^2 - \sum_{i=1}^l (\zeta_i^t - \zeta_i^p)^2}{\sum_{i=1}^l (\zeta_i^t)^2} \quad (16)$$

where l is the total number of test data, $l = 9$; ζ_i^t is the experimentally tested value, and ζ_i^p is the model prediction associated with a test. The results of the R^2 values for different test samples are shown in Figure 14.

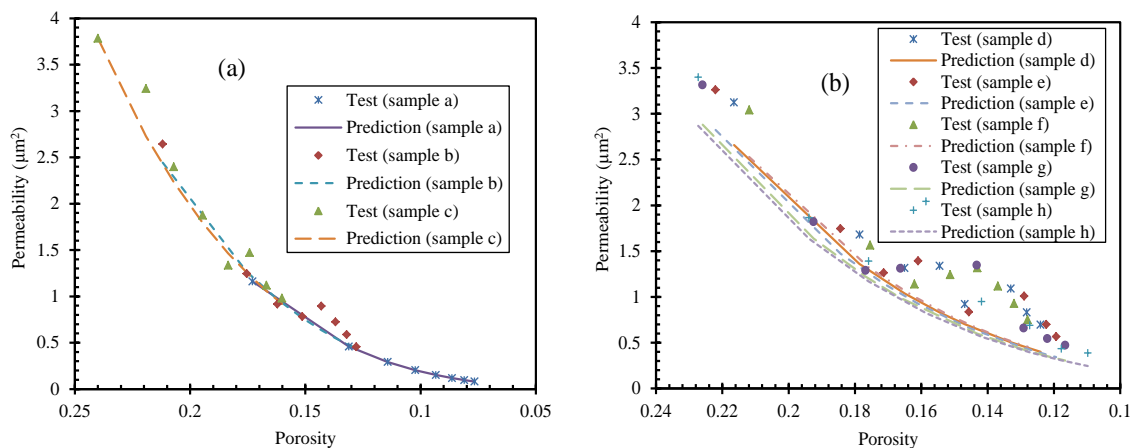


Figure 13. Permeability–porosity curves for the sample under variable conditions: (a) stress rate; and (b) original GSD.

It can be seen from Figure 13 that by using Chilingar’s empirical equation, the calculated permeability values are typically lower than those obtained from the experimental tests in this study. This could indicate that particle crushing caused a permeability increase during compression. However, as shown in Figure 14, the R^2 values in all samples were higher than 0.90, which indicates a high degree of linear correlation. Therefore, the application of effective porosity (fracture) in seepage for the prediction of permeability variations can yield sufficiently accurate results. The R^2 was almost the same for samples (a, b, and c). This is perhaps because the test time had no predictive error effect. Furthermore, a greater mass of larger diameter particles were crushed. A comparison of the predicted results corresponding to the experimental data for samples with variable GSD shows that the R^2 value

for sample (h) was the lowest (0.906), suggesting that a greater mass of larger diameter particles led to a higher proportion of crushed fines, giving greater pore volume reduction and altered permeability.

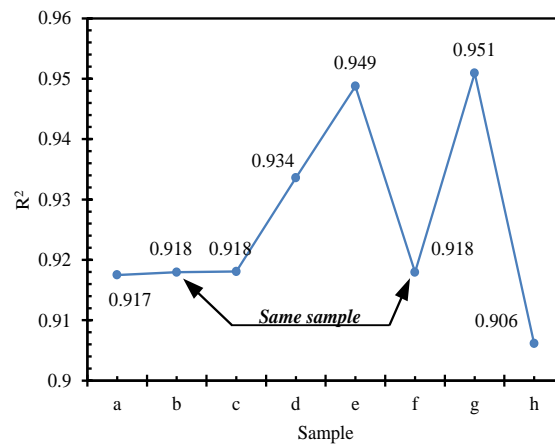


Figure 14. R^2 values for the model predictions of the permeability evolution for different samples.

5. Conclusions

In a series of experiments, the effects of applied compressive stress rates and original GSD (Figure 3) on the variations of the seepage characteristics of granular coal were investigated through experimental measurements. From the experimental observations and corresponding analyses, the following main conclusions could be drawn:

In general, gas seepage characteristics (i.e., altered porosity and altered permeability) are strongly influenced by the variable stress rate of the sample. Both porosity and permeability of granular coal decrease during compression. As samples under a higher stress state require a shorter duration of compression before fracture, the porosity and permeability decrease is incrementally smaller. Further reductions in porosity are more difficult to achieve with monotonously increased compressive stress. Samples under high stress states experienced a greater degree of particle crushing. Samples under variable stress states had variable critical compressive stress for the occurrence of isolated fractures in the larger diameter particles, after which the fracturing behavior changed from self-extending to inter-connecting. Particle attrition resulted in a significant increase in the proportion of fines and smaller diameter particles (<5 mm).

According to the experimental observations of the original GSD effect on seepage characteristics, the overall process of porosity and permeability evolution during compression can be divided into three different stages. In the first stage, the porosity and permeability decrease was mainly attributed to the reduction in pore diameter. Both the porosity and permeability decrease occurred to a lesser extent in samples with a higher mass fraction of large diameter particles. Void filling that occurred following particle attrition appeared to be the main cause for the reduction in porosity and permeability. During the second stage, the porosity and permeability in samples (d, e, f, and g) continued to decrease, and sample (h) showed the largest reduction perhaps due to rapid particle crushing (high stress state). In the third stage, consolidation of the granular coal occurred very rapidly and the porosity and permeability decreased the most for samples with a higher mass fraction of large diameter particles. An empirical model was applied to analyse the permeability and porosity evolution for the seepage and crushing processes. The prediction indicated that particle crushing may have caused the permeability increases during compression. It was hypothesized that this was also the main cause for the predictive values being lower than those obtained from the experiments in most cases. For samples under different stress states, the predictive accuracy for the porosity and permeability evolution was the same. For samples within the highest mass fraction of large diameter particles, the predictive accuracy was the lowest.

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Abbreviations

Latin Letters

h	vertical axis going through the left of sample (L)
H_i	height of granular coal during compression (L)
H_a	cylindrical tube height (L)
H_b	plunger height (L)
H_c	porous disk thickness (L)
H_d	felt pad thickness (L)
H_{ei}	height of plunger head exceeds cylindrical tube during compression (L)
m	mass of granular coal (M)
p	pore pressure ($\text{mL}^{-1} \cdot \text{T}^{-2}$)
p_{i-in}	pore pressure at intake boundary (gas in) during compression ($\text{mL}^{-1} \cdot \text{T}^{-2}$)
p_{i-out}	pore pressure at outtake boundary (gas out) during compression ($\text{mL}^{-1} \cdot \text{T}^{-2}$)
r	radius of the cylindrical tube (L)
v_{i-in}	gas seepage velocity at intake boundary (in) during compression ($\text{L} \cdot \text{T}^{-1}$)

Greek Letters

μ_g	gas viscosity ($\text{mL}^{-1} \cdot \text{T}^{-1}$)
σ_i	compressive stress ($\text{mL}^{-1} \cdot \text{T}^{-2}$)
ρ_c	coal density (mL^{-3})
ρ_g	gas density (mL^{-3})
ϕ	porosity (-)
ϕ_i	porosity during compression (-)
ζ_i	permeability of granular coal during compression (L^2)

Sub- and Superscripts

i	compressive stress state (-)
j	size range (-)

Special Symbols

∂	partial differential operator (-)
$\partial() / \partial h$	Nabla operator (L^{-1})

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