



# Article Molecular Simulation Study Based on Adsorption of Gas (CO<sub>2</sub>,O<sub>2</sub>,CH<sub>4</sub>) on Coal

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Abstract: This study aimed to further explore the adsorption properties of different gases ( $CO_2$ ,  $O_2$ , and CH<sub>4</sub>) on the coking coal surface by establishing a molecular model. Changes in the absolute adsorption capacity and the isosteric heat of adsorption of gases under different temperatures, pressures, and compositions were simulated using grand canonical Monte Carlo (GCMC) and molecular dynamics simulations. Interaction energy and energy distribution were used to analyze the adsorption behavior of gases, and the diffusion properties were investigated using the diffusion coefficient and diffusion activation energy. The absolute adsorption results fit well with the Langmuir-Freundlich model. The absolute adsorption capacity had a significant positive correlation with pressure and the corresponding mole fraction, and a significant negative correlation with temperature. The competitiveness, based on binary adsorption selectivity, was in the order of  $CO_2 > O_2 > CH_4$ . The isosteric heat of adsorption of  $CH_4$  was slightly higher than that of  $O_2$ , and that of  $CO_2$  was 1.49–1.64 times that of  $O_2$  and  $CH_4$ . The isosteric heat of the adsorption of gases was also barely influenced by temperature and pressure. The interaction energy between CO<sub>2</sub> and coal was greater than that of  $O_2$  or  $CH_4$ , but the high pressure and high content were not conducive to the adsorption of  $O_2$  by  $CO_2$ . The preferred adsorption site for  $CO_2$  was stronger than that for  $O_2$  and  $CH_4$ , and its peak value negatively correlated with the molar fraction. The diffusion coefficient for single component gases initially increased and then decreased with increased pressure, showing a positive correlation with temperature. A close inverse correlation existed between diffusion activation energy and pressure. These results revealed the microscopic adsorption and diffusion regularities of CO<sub>2</sub>, O<sub>2</sub>, and CH<sub>4</sub> in the coal model, indicating great significance in accurately predicting coal fires.

Keywords: coal spontaneous combustion; competitive adsorption; molecular simulation

# 1. Introduction

Under the influence of the national characteristics of "poor oil, less gas, and rich coal", coal dominates the energy system in China [1]. However, the complex geohydrological structure of mines and other disaster-inducing factors have conferred 56% of recoverable coal seams with self-combustion tendency [2], of which about 49% are affected by combined disasters [3]. Thus, preventing and controlling coal's spontaneous combustion has become a research hotspot in coal mine safety [4].

The initiation of coal autoignition is an extremely intricate process of coal–oxygen physicochemical adsorption. The coal first physically adsorbs oxygen to saturation, at which point the functional groups on the coal surface chemically react with the oxygen to change the molecular structure and release heat. Heat build-up leads to an increase in temperature, triggering coal's spontaneous combustion [5]. Coal–oxygen adsorption provides the early basis for coal self-ignition, a process of significant importance in revealing the theory of coal spontaneous combustion. However, the influence of  $CH_4$  and  $CO_2$  in the coal seam on coal–oxygen adsorption cannot be ignored. To this end, scholars have used molecular simulations to investigate the competitive adsorption regularity for  $CO_2$  and  $CH_4$ 



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**Copyright:** © 2023 by the authors. Licensee MDPI, Basel, Switzerland. This article is an open access article distributed under the terms and conditions of the Creative Commons Attribution (CC BY) license (https:// creativecommons.org/licenses/by/ 4.0/). in coal. Lu et al. [6] studied the physisorption process of oxygen by functional groups in coal. Their results show that physisorption begins to change into chemisorption as the adsorption amount increases. Zhou et al. [7], Yang et al. [8], Zhang et al. [9], and Ren et al. [10] used a coal macromolecular model to adsorb CH<sub>4</sub> and CO<sub>2</sub> and found that the adsorption capacity of CO<sub>2</sub> was stronger than that of CH<sub>4</sub> [11]. Yu et al. [12], Liu X et al. [13], and Dang et al. [14] investigated the impact of oxygenic groups on the competitive adsorption of  $CO_2$ ,  $CH_4$ , and N<sub>2</sub> in coal using GCMC and density functional theory (DFT) simulations. They found that the strong quadrupole moment and polarization capacity of CO<sub>2</sub> makes it more selectively adsorbed by functional groups than  $CH_4$  and  $N_2$ . Cheng et al. [15], Xiang et al. [16], and Ding et al. [17] revealed the different adsorption and diffusion mechanisms of CH<sub>4</sub>, CO<sub>2</sub>, and N<sub>2</sub> gases from three aspects: adsorption isotherms, adsorption heat, and diffusion coefficients using molecular simulation and quantum chemistry methods. They showed that the order of adsorption capacity for the three gases is  $CO_2 > CH_4 > N_2$ , whereas the diffusion capacity follows the opposite sequence. Yu et al. [18] and Tang et al. [19] conducted experimental studies on the competitive adsorption of mixed gases under different volume fractions. They found that the adsorption amount in multicomponent gases was affected by the adsorption power, molecular properties, and the partial pressure. Wang et al. [20] and Zhang et al. [21] studied the adsorption of  $CH_4$ ,  $CO_2$ ,  $N_2$ ,  $O_2$ , and their mixed gases on a coal surface model. The presence of  $CO_2$  was found to greatly reduce the adsorption amount of  $CH_4$  in the binary system, and the adsorption of oxygen was more likely to be affected by the volume fraction of methane.

In summarizing the competitive adsorption among multicomponent gases in coal, previous studies have primarily focused on improving the efficiency of CBM extraction, and most of them made comparisons with N<sub>2</sub>. Only a few studies have investigated the influence of the competitive adsorption behaviors of  $CH_4$ ,  $CO_2$ , and  $O_2$  on coal's spontaneous combustion. However,  $O_2$  is an essential gas for the spontaneous combustion of coal, so the study of its competitive relationship with  $CH_4$  and  $CO_2$  in coal is an essential prerequisite for ensuring the safe operation of coal mines and preventing the occurrence of fire accidents due to the spontaneous combustion of coal. Therefore, taking  $CO_2$ ,  $O_2$ , and  $CH_4$  as the research objects, the present study used GCMC to simulate and analyze the adsorption behaviors of these gases in the molecular structure of coking coal. This research aimed to provide a theoretical basis for coal mine fire prevention at the microscopic level.

# 2. Construction and Simulation Method of the Coal Molecular Model

#### 2.1. Coal Molecular Configuration

The coking coal was obtained from the 12th Coal Mine of Pingdingshan, Henan Province. For this research, elemental and industrial tests were used to analyze the coking coal, and the results are shown in Table 1. According to <sup>13</sup>C-NMR, XRD, XPS, and other experiments, we obtained information on the distribution of hydrocarbon atoms, the arrangement of aromatic structures, and the presence of functional groups containing nitrogen, oxygen, and sulfur. By combining the results of the industrial analysis with the results of the elemental analysis, the molecular formula of coking coal was determined to be  $C_{209}H_{140}O_{17}N_4$ , which was derived from modeling methods described in the literature [22]. The two-dimensional structure diagram and the structural parameters are displayed in Figure 1 and Table 2, respectively.

**Table 1.** The basic parameters of the coal sample.

Proximate Analysis (%)					Ultim	ate Analys	sis (%)	
M <sub>ad</sub>	A <sub>ad</sub>	$\mathbf{V}_{daf}$	FC <sub>ad</sub>	С	Н	0	Ν	S
1.68	22.64	17.3	62.58	84.02	4.70	9.33	1.88	0.07

	<b>NØ 1 1 147 * 1</b> 7	elecular Weight Element Content (				
Molecular	Molecular Weight	С	Н	0	Ν	
C <sub>209</sub> H <sub>140</sub> O <sub>17</sub> N <sub>4</sub>	2982	84.02	4.7	9.1	1.8	
H <sub>3</sub> C () ()						

Table 2. The structural parameters of the coking coal.

Figure 1. The molecular configuration diagram of coal (two-dimensional).

#### 2.2. Optimization of the Coal Macromolecule Model

Since only the structural model with the lowest energy represented the optimal configuration under study, the two-dimensional structure of coking coal was imported into Materials Studio software and the Forcite module was used to calculate MM and MD. The main purpose of MD calculation was to avoid excessive calculation of adsorption simulation due to the complex macromolecular structure. The MM parameters were [23] that the calculation method was a smart minimizer, the maximum number of iteration steps was set to 5000 steps, and the charge distribution and the force field were the charge equilibrium method and Dreiding. The electrostatic and Van der Waals values were calculated using the atom-based method. The MM calculation overcomes the disadvantage of only the local minimum value being obtained via MD. The minimum value of the whole potential energy surface was found using the annealing dynamics simulation. The MD parameters were the following [24]: the NVT ensemble (constant particle number N, volume V, and temperature T in the simulation system remain unchanged) was selected, the temperature range was 300–600 K, and it was heated up 60 K each time and cycled ten times. The temperature control program selected Nose, and the step size was 1 fs. The MM calculations were performed on the output configuration at the end of each cycle, and the parameters were set as described above. The optimized results of MM and MD are shown in Figure 2a,b, respectively. The adsorbent configuration after molecular optimization is shown in Figure 3.

Density simulation not only added periodic boundary condition to coal molecules, but also explored the optimal configuration under periodic boundary conditions. Finally, the reliability of the modeling method was verified by comparing the cell density obtained after passing the minimum energy point with the actual coal sample density. The periodic boundary conditions were added to the model using the Amorphous Cell module. The following parameters were used to simulate the density [25]: the calculation used Medium, the Dreiding was used to force field, and the charge was calculated using the charge balance method (QEq). The initial density was 0.6 g/cm<sup>3</sup>, the final density was 1.8 g/cm<sup>3</sup>, and the interval was 0.05 g/cm<sup>3</sup>. By analyzing the relationship between density and potential energy, it was found that the structure had the lowest potential energy when the density was 1.35 g/cm<sup>3</sup>. The lattice parameters of the optimal configuration are  $L_x = L_y = L_z = 1.3912$  nm,  $\alpha = \beta = \gamma = 90^\circ$ , as shown in Figure 2c.



Figure 2. (a) geometry optimization, (b) annealing treatment, (c) cubic cell model, (d) connelly surface.



**Figure 3.** The optimized molecular structure of an adsorbent: (a)  $CH_4$ , (b)  $CO_2$ , and (c)  $O_2$ .

After the NPT (P = 1 MPa, T = 298 K) ensemble simulation, the final density of the model was stabilized at 1.39 g/cm<sup>3</sup>, which was close to the measured value. The effective pore volume and surface area for the model were determined using the Atom Volumes and Surfaces tool, and were 2414 m<sup>2</sup>/g and 0.0209 cm<sup>3</sup>/g. The Connery surface of the model [26] is shown in Figure 2d.

To closely approach the real porous state of coal, a supercell molecule comprising  $2 \times 2 \times 2$  original molecules of coking coal was constructed to form pores of different sizes [27]. This paper used the GCMC method to simulate the relationship between the adsorption capacity of CO<sub>2</sub>, O<sub>2</sub>, and CH<sub>4</sub> on coal surface and fugacity, wherein fugacity was converted from pressure using the Peng–Robinson equation [28]. The adsorption characteristics of gas molecules on the coal surface were investigated using the adsorption module to obtain the adsorption isotherm, adsorption site, action energy, diffusion coefficient, and adsorption selectivity. The parameters of the sorption module were set as follows [29]: the task item was Fix Pressure and the precision was Customized. The Metropolis method was used to

calculate the energy change in adsorbed gas molecules on the surface of the coal molecular model. The number of molecules N in the equilibrium state of the simulated system was obtained using the  $\mu$ VT ensemble (chemical potential  $\mu$ , volume V, and temperature T of the model system remained unchanged). To achieve thermodynamic equilibrium and save on computation time, the total number of simulation steps was determined to be  $2 \times 10^7$  Monte Carlo steps. The number of equilibrium steps was set as  $1 \times 10^7$  to ensure adsorption equilibrium and the number of production steps was set as  $1 \times 10^7$  to sample the exact average date. The models obtained at different temperatures and pressures using the Fix pressure method were taken as the initial model, which was optimized using MD simulation. The Dynamic task was used to calculate the kinetic coefficients, such as diffusion coefficient and diffusion activation energy.

## 3. Simulation Result Analysis

# 3.1. Absolute Adsorption Capacity

#### 3.1.1. Single Component Gas Adsorption Capacity

The simulation data were fitted using Langmuir, Freundlich, and Langmuir–Freundlich models [30]. The Langmuir–Freundlich model could describe the entire adsorption for gases very well, and its fitting formula [31] was as follows:

$$q = \frac{abp^{1/n}}{1 + bp^{1/n}}$$

where *q* is the adsorption amount (mL/g), *a* is the gas maximum single layer adsorption amount, *p* is the adsorption pressure (MPa), *b* is the adsorption equilibrium constant (MPa<sup>-1</sup>), and *n* is the surface heterogeneity of the adsorbent. The fitting curves of the adsorption isotherm are shown in Figure 4.



Figure 4. The adsorption isotherm and Langmuir–Freundlich fits of (a) CO<sub>2</sub>, (b) O<sub>2</sub>, and (c) CH<sub>4</sub>.

With increased pressure, the adsorption isotherm for  $CO_2$  initially increased quickly and remained stable,  $O_2$  increased slowly at the low pressure stage and then gradually became saturated, and  $CH_4$  increased linearly. These phenomena not only showed that there was a critical value of the influence of pressure on the adsorption capacity [32], but the molecular weight of the gas also affected the adsorption amount because coal molecules had a limited effective adsorption point. Therefore,  $CO_2$  had a high adsorption amount due to its greater molecular weight, compared to  $CH_4$  and  $O_2$ . In comparison with the simulation results of Qiang [33], the absolute adsorption isotherm for  $CO_2$  in this work was found to be qualitatively in agreement, but quantitatively higher. The reason was that the hydrogen, oxygen, and nitrogen content in coking coal was higher, indicating that there were more hydroxyl, carboxyl, and methoxyl groups interacting with  $CO_2$ . The relationship between the adsorption amount of gases under the same condition was  $CO_2 > O_2 > CH_4$ . The interaction energy between gases and coal molecules was an important basis for sequencing. The adsorption capacity decreased with the temperature increase. This was because the initial stage of adsorption was physisorption caused by intermolecular suction, and the increase in temperature reduced the original weak binding force and led to desorption.

The fitting parameters and correlation coefficient of the Langmuir–Freundlich models are summarized in Table 3. All R<sup>2</sup> values exceeded 0.98, confirming the reliability of the Langmuir–Freundlich model. The relationship of the parameter was  $a_{(CO2)} > a_{(O2)} > a_{(CH4)}$ , which means that CO<sub>2</sub> was the first gas to attain stability and had the maximum adsorption capacity. The parameter b was inversely proportional to the pressure required for saturation, indicating that the magnitude of the pressure required by the gas was CH<sub>4</sub> > O<sub>2</sub> > CO<sub>2</sub>, so that the adsorption rate for CO<sub>2</sub> increased the fastest under a low pressure [34]. With increased temperature, the fitted parameters gradually decreased. This suggests that the temperature was not favorable to gas adsorption, but provided energy for the gas to escape from the coal surface.

Table 3. The Langmuir–Freundlich fitting parameters.

Tommorphuse (V)	CO <sub>2</sub>				O <sub>2</sub>			$CH_4$				
Temperature (K)	а	b	n	<b>R</b> <sup>2</sup>	а	b	n	<b>R</b> <sup>2</sup>	а	b	n	R <sup>2</sup>
288	5.910	1.981	0.881	0.996	5.818	1.417	2.853	0.98	4.671	0.178	3.707	0.999
298	5.793	0.891	0.435	0.985	4.491	0.668	2.005	0.988	2.798	0.181	2.587	0.998
308	5.653	0.831	0.316	0.986	4.424	0.664	1.957	0.989	1.59	0.06	1.35	0.992
318	5.581	0.018	0.010	0.984	4.625	0.099	0.216	0.98	1.146	0.035	0.261	0.996

## 3.1.2. Multi-Component Gas Adsorption Capacity

To investigate the adsorption capacity between different gases, the adsorption isotherms for multi-component gases on coal at 298 K were simulated. To analyze the effect of molar fraction on adsorption behavior in the binary component system and the difference in competitive adsorption of gases in the ternary component system, the component ratios were set to 1:4, 2:3, 3:2, 4:1, and 1:1:1. The simulation results are shown in Figure 5a–d.



Figure 5. Absolute adsorption isotherm of binary (a-c) and ternary (d) systems.

Figure 5 shows that the adsorption amount of multi-component gas increased with increased molar fraction. In the  $CO_2/O_2$  and  $CO_2/CH_4$  binary system, the adsorption amounts of  $CO_2$  were always greater than those of  $O_2$  and  $CH_4$  at different molar fractions, manifesting that  $CO_2$  had a more competitive capacity than  $O_2$  and  $CH_4$ . A comparison of Figure 5a,b revealed that the maximum values of  $CO_2$  in the  $CO_2/O_2$  and  $CO_2/CH_4$  systems were 5.32 and 7.57 mmol/g, respectively, demonstrating that  $O_2$  was more competitive than  $CH_4$  and consistent with Figure 5c. By comparing the adsorption quantities within the pure-gas system and the  $CO_2/O_2/CH_4$  system, it could be found that the adsorption amount for  $O_2$  dramatically decreased. It has been shown that the presence of  $CO_2$  affected the physical adsorption of  $O_2$  in coal and reduced the possibility of the spontaneous combustion of coal from the source.

## 3.1.3. Adsorptive Selectivity

To further research the preferential adsorption ability of the gases on coal, adsorption selectivity was used to describe the competition hierarchy for multicomponent gas.  $S_{m/n}$  could be defined as follows:

$$S_{m/n} = \frac{x_m/x_n}{y_m/y_n}$$

where  $x_m$  (or  $x_n$ ) and  $y_m$  (or  $y_n$ ) are the mole fraction of species m (or n) in the adsorbed phase and bulk phase, respectively. The adsorption selectivity was larger than 1, indicating that the competitive adsorption of adsorbate m in the multi-component was stronger than that of adsorbate n, and greater selectivity corresponded with stronger adsorption.

 $S_{CO_2/O_2}$  decreased with the increased pressure and molar fraction of CO<sub>2</sub>, as shown in Figure 6a, indicating that high pressure and high content reduced the competitiveness of CO<sub>2</sub>. This finding was primarily due to the CO<sub>2</sub> saturation being reached during the high-pressure phase, whereas O<sub>2</sub> was always on the rise. Figure 6b showed that  $S_{CO_2/CH_4}$ was inversely proportional to pressure and positively proportional to the molar fraction of CO<sub>2</sub>, indicating that a higher CO<sub>2</sub> content corresponded with a stronger the competitive adsorption. As shown in Figure 6c,  $S_{O_2/CH_4}$  was proportional to the pressure and the molar fraction of O<sub>2</sub> because the amount and rate of adsorption for O<sub>2</sub> was far beyond that of CH<sub>4</sub>. The analysis of adsorption selectivity revealed that the competitiveness remained at CO<sub>2</sub> > O<sub>2</sub> > CH<sub>4</sub>, and this conclusion was confirmed by the tri-component system shown in Figure 6d.

## 3.2. Isosteric Heat of Adsorption

### 3.2.1. Single Component Gas Adsorption Heat

To some extent, the adsorption capacity could be reflected by the magnitude of the isosteric heat of adsorption. A stronger interaction energy between the gas and coal corresponded with greater isosteric heat [35]. The relationship between the adsorption heat and pressure of CO<sub>2</sub>, O<sub>2</sub>, and CH<sub>4</sub> at different temperatures is shown in Figure 7. The adsorption heat of each gas was only slightly affected by temperature and pressure. The adsorption heat of CO<sub>2</sub> initially decreased and then increased at around 8.625 kcal/mol. This finding may be due to the energy of the adsorbate-adsorbent interaction dominating at low pressure, whereas the adsorbate-adsorbate interaction contributed more at high pressure. Conversely, the isosteric heat for  $O_2$  and  $CH_4$  showed an overall decreasing trend of about 0.5–1 kcal/mol, indicating that adsorption was dominated by adsorbate– adsorbent interaction. The isosteric heat of CH4 was a little superior to that of O2, and that of CO<sub>2</sub> (8.44–8.73 kcal/mol) was about 1.49–1.56 and 1.55–1.64 times that of O<sub>2</sub> and CH<sub>4</sub>, respectively. Thus, the order of adsorption heat was  $CO_2 > CH_4 > O_2$ , inconsistent with the order of the adsorption amount. This indicated that the isosteric heat of adsorption was only one of many factors affecting the adsorption capacity, which reflected the adsorption capacity to a certain extent. Because the heat levels of  $CO_2$ ,  $O_2$ , and  $CH_4$  were less than 10 kcal/mol, the adsorption on the coal surface was a physical process [36].



Figure 6. Adsorption selectivity of binary (a-c) and ternary (d) systems.



Figure 7. The isosteric adsorption heat of single component gas.

Henry's law of adsorption describes the linear relationship between adsorption amount and equilibrium pressure at low pressure, and the formula was as follows:

$$n = K_H P$$

where *n* is the adsorbing capacity,  $K_H$  is the Henry constant, and *P* is the adsorption pressure. The Henry constant could characterize the adsorbate affinity, and it decreased with decreased  $K_H$ . The relationship of adsorption affinity for three gases was  $CO_2 > O_2$ >  $CH_4$ , as shown in Figure 8. This result was due to the permanent quadrupole moment of the  $CO_2$  molecule, which created a stronger electrostatic force on the surface of coal molecules [37]. The negative correlation showed that the affinity of these gases could be reduced by the creasing temperature. The  $K_H$  values of  $CO_2$  were more sensitive to temperature changes, and that of  $CH_4$  slowly decreased with increased temperature.



Figure 8. The Henry constant for single component gas.

## 3.2.2. Multi-Component Gas Adsorption Heat

The isosteric heat of adsorption in multi-component systems is shown in Figure 9, and it depended primarily on gas species and the system associated with it. Figure 9a shows that the adsorption heat for  $CO_2$  was proportional to its molar coefficient in the  $CO_2/CH_4$  system, but inversely proportional in the  $CO_2/O_2$  system, consistent with the trend of adsorption selectivity in binary systems. Influenced by the competition between  $CO_2$  and  $CH_4$ , the adsorption heat of  $O_2$  in the  $CO_2/O_2$  and  $O_2/CH_4$  systems differed. The adsorption heat of O2 decreased with increased molar fraction, which was contrary to that of CH<sub>4</sub>, indicating that the high CH<sub>4</sub> content was more competitive than that of  $O_2$ . By comparison with Figure 9b and c, it could be seen that the adsorption heat of  $O_2$  and  $CH_4$ in the  $O_2/CH_4$  system followed the same trend as that of pure gas, whereas it fluctuated more when CO<sub>2</sub> was involved. This phenomenon was due to the adsorption heat of pure  $CO_2$ , showing a local minimum that could be found only in strongly adsorbed gas [38]. The order of adsorption heat of gases in the  $CO_2/O_2/CH_4$  system was compatible with that of pure gases. The change trends for  $O_2$  and  $CH_4$  showed more obvious fluctuations than those of  $CO_2$ , as shown in Figure 9d. This finding indicated a strong competition between  $O_2$  and  $CH_4$  for adsorption heat.

## 3.3. Interaction Energy

The interaction energies of  $CO_2$ ,  $O_2$ , and  $CH_4$ , including van der Waals energy and electrostatic energy, were analyzed at different molar fractions in multi-component systems to further investigate the effect of competitive adsorption of gases for the interaction energy [39]. The results are shown in Figure 10.

As shown in Figure 10a,b, the van der Waals energy accounted for more than 79% and 77% of the total interaction energy in the  $CO_2/O_2$  and  $CO_2/CH_4$  systems, respectively, whereas the rest of the electrostatic energy originated from  $CO_2$ . This was because  $CO_2$  was electrically charged and generated electrostatic energy when adsorbed. Meanwhile, the higher molar fraction for  $CO_2$  corresponded with stronger interaction energy, indicating that it was more likely to adsorb than  $O_2$  and  $CH_4$ , and the adsorption system was more stable. For the  $CO_2/O_2$  system, the interaction energy was slightly reduced when the pressure exceeded 8 MPa and the molar fraction of  $CO_2$  exceeded 60%, demonstrating that the high pressure and high content reduced the competitiveness of  $CO_2$ . For  $O_2/CH_4$  binary systems, up to 98% of the total energy of interaction was van der Waals energy, and only a small amount of energy originated from electrostatic energy. The interaction energy was proportional to the pressure and the molar fraction of  $O_2$ , indicating that  $O_2$  was more stable than  $CH_4$ . The results showed that the relationship of the adsorption stability of the gases was  $CO_2 > O_2 > CH_4$ , which was consistent with the relationship of adsorption amount.

Figure 9. The isosteric heat of adsorption of  $CO_2$ ,  $O_2$ , and  $CH_4$  in binary (**a**-**c**) and ternary (**d**) systems.



Figure 10. Interaction energy: (a) in the  $CO_2/O_2$  system; (b) in the  $CO_2/CH_4$  system; (c) in the  $O_2/CH_4$  system; and (d) the total energy and adsorption amount in the  $CO_2/O_2/CH_4$  system.





As shown in Figure 10d, the interaction energy in the  $CO_2/O_2/CH_4$  system decreased from -2.962 kcal/mol to -122.163 kcal/mol with the pressure from 0.1 MPa to 10 MPa. The van der Waals energy accounted for more than 73%. The larger the absolute value of the interaction energy, the more prone it was to adsorb. The van der Waals energy, electrostatic energy, and total energy initially increased rapidly and then slowly with increased pressure, which was consistent with the increasing trend of adsorption capacity under the  $CO_2/O_2/CH_4$  system [40].

# 3.4. Energy Distribution

The adsorbed sites could be identified using the energy distribution and be used to analyze competitive adsorption. Larger negative values indicated a stronger interaction energy and more favorable adsorption sites [41]. The energy distribution of pure  $CO_2$ ,  $O_2$ , and  $CH_4$  at different temperatures was shown in Figure 11a. The preferential adsorption site for  $CO_2$  was lower than  $O_2$  or  $CH_4$ , and the peak for  $O_2$  was almost equal to that for CH<sub>4</sub>. This was consistent with the order of equal heat of adsorption. The distinct potential energy peak was around -10.5 kcal/mol, corresponding with favorable adsorption sites for  $CO_2$ . Its peak decreased with increased temperature, which was because the temperature stimulated the activity of  $CO_2$  molecules, so that the adsorbed gas molecules began to diffuse. Another peak at around -6.5 kcal/mol represented the favorable adsorption sites for  $O_2$  and  $CH_4$ . The peaks of the preferential adsorption sites for  $O_2$  and  $CH_4$  moved toward the lower energy region with increased pressure, and a new peak at -4 kcal/mol formed at the secondary adsorption site. This finding was primarily due to the increase in adsorption volume caused by pressure, so that the priority adsorption sites gradually became saturated. Then, a large number of gas molecules shifted to the secondary adsorption sites, resulting in a movement in the relative importance of the adsorption sites.



Figure 11. Cont.



**Figure 11.** Energy distribution: (**a**,**b**) in a single component adsorption system; (**c**,**d**) in different adsorption systems.

The energy distribution of gases in the multi-component systems was compared at different molar fractions in order to study competition among adsorption sites, as shown in Figure 11b–d. In the  $CO_2/O_2$  system, the peak value of the preferential adsorption site increased with increased molar fraction, implying that  $CO_2$  and  $O_2$  gradually reached saturation with increased adsorption capacity. The potential energy shifted to the right and the peak decreased with increased molar fraction in the  $CO_2/CH_4$  system, showing that the high content did not facilitate the competitive adsorption of gas molecules at the preferred site. By comparing the energy–distribution curve of  $O_2$  in the  $CO_2/O_2$  and  $O_2/CH_4$  systems, we found that adding  $CH_4$  increased the peak at the second adsorption site. This finding indicated that the presence of  $CH_4$  forced the  $O_2$  molecules to diffuse away from the preferred adsorption site, thereby inhibiting  $O_2$  from reaching saturation. The molecular proportion of the preferred adsorption site for  $O_2$  (-5 kcal/mol) in the binary components was reduced.

## 3.5. Diffusion Property

When gas molecules made contact with the coal surface, different pressures and concentration gradients were formed by different adsorption capacities. The gas diffused from high to low under the gradient, which follows the microscopic principle of diffusion [42]. The self-diffusion coefficients of  $CO_2$ ,  $O_2$ , and  $CH_4$  were calculated using MD simulation to reveal the diffusion regularity of the gas on coal [43]. From the Einstein diffusion equation, the  $D_s$  could be calculated as follows [44]:

$$D_{s} = \frac{1}{6N} \lim_{t \to \infty} \frac{d}{dt} \left\{ \sum_{i=1}^{N} [r_{i}(t) - r_{i}(0)] \right\}^{2} = \frac{k_{MSD}}{6}$$

where  $K_{MSD}$  is the slope of the fitting curve of MSD, and  $MSD(t) = \sum_{i=1}^{N} \langle |r_i(t) - r_i(0)|^2 \rangle$ ; *N* is the number of gas molecules;  $r_i(t)$  is the Cartesian position vector of gas molecule in the microcrystallite at the time *t*; and  $r_i(0)$  is the initial position vector of the gas molecule. The diffusion coefficient and diffusion activation energy for gases on coal at different pressures are depicted in Figure 12.



**Figure 12.** The self-diffusion coefficient  $(\mathbf{a}-\mathbf{c})$  and the diffusion activation energy  $(\mathbf{d})$  of CO<sub>2</sub>, O<sub>2</sub>, and CH<sub>4</sub>.

The self-diffusion coefficient of gases was positively correlated with temperature because the higher temperature corresponded with the higher internal energy of the molecules. According to the law of conserving energy, the internal energy between molecules could be converted into kinetic energy, thereby intensifying the movement of molecules and making it easier for them to diffuse. The relationship between the self-diffusion coefficients of the gas was  $CO_2 > O_2 > CH_4$  at the same temperature, consistent with the results of Kelemen et al. [45]. The self-diffusion coefficient initially increased and then decreased with pressure. The main reason for the reduction was that gas molecules stacked up more tightly and interacted more strongly under high pressure.

The diffusion of gas in coal molecules was an activation process. The diffusion activation energy could be reckoned using the Arrhenius equation [46], and the specific formula is as follows:

$$D = D_0 \exp\left(-\frac{E_a}{RT}\right)$$

1

where  $D_0$  is the pre-exponential factor;  $E_a$  is the apparent activation energy, kcal/mol; R is the ideal gas constant; and T is the temperature, K. InD is fitted well to the reciprocal of temperature, and the calculated result is listed in Table 4. The activation energy of gas diffusion was found to be negatively correlated with pressure, and the activation energy of  $O_2$  was approximately twice that of  $CO_2$ .

**Table 4.** The diffusion activation energy of CO<sub>2</sub>, O<sub>2</sub>, and CH<sub>4</sub>.

Pressure	CO <sub>2</sub>	O <sub>2</sub>	CH <sub>4</sub>
0.1 MPa	19.65	39.60	32.29
1 MPa	17.25	35.83	27.12
10 MPa	10.83	29.05	27.73

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## 4. Conclusions

To research the microcosm mechanism of  $CO_2$ ,  $O_2$ , and  $CH_4$  adsorption and diffusion on coal, a realistic macromolecular coal model was established. GCMC and MD molecular simulations were performed in single, binary, and ternary systems, considering the effect of temperature, pressure, and molar fraction.

(1) Adsorption isotherms were well fitted with the Langmuir–Freundlich model. The absolute adsorption amount was directly proportional to the pressure and inversely to the temperature. The adsorption of multi-component gases showed that adsorption amount was proportional to the molar fraction, but high pressure and high content reduced the competitiveness for CO<sub>2</sub>. The competitive capacities were  $CO_2 > O_2 > CH_4$ , based on adsorption selectivity. By comparing the adsorption amount of  $O_2$  under different component systems, we found that  $CO_2$  significantly reduced the adsorption amount of  $O_2$ .

(2) The isosteric heat of adsorption of CO<sub>2</sub> (8.44–8.73 kcal/mol) was much greater than that of O<sub>2</sub> or CH<sub>4</sub> (5.12–5.83 kcal/mol). The difference in order between the adsorption quantity and the adsorption heat for the three gases meant that the adsorption amount was influenced by the adsorption heat, and many other factors. The adsorption heat was affected by the molar fraction and competition from other gases in mixed adsorption systems, which changed the adsorption sites and adsorption spaces and influenced the interaction energy. The presence of CO<sub>2</sub> affected the trend of the equivalent heat of adsorption of another gas with which it competed, and the existence of CH<sub>4</sub> caused large fluctuations in the adsorption heat of O<sub>2</sub>.

(3) The electrostatic energy and high van der Waals energy between CO<sub>2</sub> and coal resulted in an interaction greater than with O<sub>2</sub> and CH<sub>4</sub>. The greater interaction energy corresponded with the greater adsorption amount. In a competitive adsorption system, CO<sub>2</sub> and CH<sub>4</sub> changed the relative importance of the competitive adsorption sites for O<sub>2</sub>, thereby inhibiting O<sub>2</sub> adsorption. The gas diffusion coefficient was inversely correlated with the temperature under the same pressure. The diffusion coefficient increased and then decreased with increased pressure at the same temperature. The order of diffusion activation energy was O<sub>2</sub> > CH<sub>4</sub> > CO<sub>2</sub>, which was negatively correlated with pressure.

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