Pyrolytic Characteristics and Kinetics of Fractional Step Demineralized Coals

MEIJUN WANG, CHUNHUI FU, XIURONG REN, LIPING CHANG* and KECHANG XIE

State Key Laboratory Breeding Base of Coal Science and Technology Co-founded by Shanxi Province and the Ministry of Science and Technology, Taiyuan University of Technology, Shanxi Taiyuan 030024, P.R. China

*Corresponding author: Tel/Fax: +86 351 6010482; E-mail: lpchang@tyut.edu.cn; nmgwmj1985@163.com

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Two Chinese coals with different reductive properties and mineral compositions were selected as the experimental samples to determine their pyrolytic characteristics. The minerals in two coals were successively leached with water, HCl and HF solutions to investigate the influence of minerals on the pyrolytic properties of the two coals. The results show that the alkali and alkaline earth metals (AAEM) minerals play an obvious catalytic role on the pyrolysis of different reductive coals, in particular, the secondary devolatilization phase. The pyrolysis process of all coal samples studied can be described by three independent first order kinetic model and the kinetic parameters of these samples during pyrolysis are affected by the minerals removed in different leaching step.

Key Words: Coal, Fractional step demineralization, Pyrolysis, Kinetics.

INTRODUCTION

The existence of inherent minerals in coal strongly depends on the geological location of the coal seam. The most common minerals in coal are quartz, clay minerals, feldspars and carbonates such as siderite, calcite and dolomite, as well as sulfide minerals such as pyrite^{1,2}. These inherent minerals exert a substantial influence on the technological processes associated with the heat treatment of coals³⁻⁵. They may play a catalytic, restraining or inert role on the major coal utilization processes, such as hydrogenation⁶, gasification⁷, combustion⁸ and pyrolysis⁴, depending on their content and chemical composition, which vary from coal to coal.

Weakly reductive coals are the main coal resources in Chinese western region, which are formed in inland gathering environment and subjected to a strong oxidative. These weakly reductive coals will play a vital role in the energy supply of China. Compared with the typical reductive coal, they have the characteristics of typically weak reducibility, such as low ash, sulfur and phosphorus contents but high inertinite, oxygen content, calorific value and aromatic structure⁹; whilst they have lower-level silicon and aluminum, higher-level alkali metals, alkaline earth metals and iron content. Based on the specific characteristic, these weakly reductive coals show a different performance in thermal conversion compared to the reductive coal⁹⁻¹¹.

In this work, a weakly reductive coal from Shendong coal mine in the western region and a reductive coal from Pingshuo

coal mine in North China were selected and investigated. This work was aimed to clarify the roles of the minerals on the pyrolysis properties of these two varieties of coals. The water-leaching, HCl-leaching and HF-leaching methods were used to remove the different types of minerals. Pyrolytic characteristics of raw coals and fractional step demineralized coals were systematically investigated using thermal gravimetric analysis, the kinetic analysis of coal pyrolysis can be determined to further reveal the effect of inherent minerals on the coal pyrolysis.

EXPERIMENTAL

Sample preparation: Two coal samples from the Pingshuo coal and Shendong coal mines were crushed, ground to the particle sizes range of 0.125-0.154 mm and preserved in a closed vessel away from light and oxygen. The ultimate and proximate analyses of these samples are presented in Table-1. The ash composition analyses of raw coals are presented in Table-2.

The minerals in two coals were removed using a fractional step leaching method in water, HCl and HF solutions, respectively. For water leaching process, a 10 g coal sample was mixed with 100 mL ultrapure water. The slurry was stirred for 24 h at room temperature, filtered and dried at 80 °C in the vacuum oven for 24 h. For HCl leaching process, the 10 g water-leached coal was mixed with 60 mL HCl solution (36 wt %, HCl:H₂O = 1:1, vol %). The slurry was stirred for 24 h at room temperature, filtered, leached with ultrapure water

TABLE-1
PROXIMATE AND ULTIMATE ANALYSES OF COAL SAMPLES USED IN EXPERIMENT

Sample	Proximate analysis (wt%)			Ultimate analysis (wt%, daf)				
	M _{ad}	A _d	V _{daf}	C	H	O*	N	S
PSR	2.2	17.9	37.2	80.4	5.2	11.9	1.4	1.1
PSW	1.7	15.2	29.4	79.9	4.7	13.2	1.3	0.9
PSM	2.0	12.7	30.7	81.2	4.8	11.8	1.3	0.9
PSD	0.7	1.1	32.2	81.9	4.7	11.3	1.2	0.9
SDR	9.8	4.5	33.7	79.8	4.2	14.6	0.9	0.5
SDW	5.2	3.6	34.5	79.5	4.5	14.7	0.9	0.4
SDM	4.4	2.4	34.5	78.6	4.3	15.7	0.8	0.6
SDD	0.7	0.5	32.8	76.8	4.7	17.1	1.1	0.3

Note: ad = air-dried basis; d = dry basis; daf = dry and ash-free basis; * by difference.

PSR = Pingshuo raw coal sample; PSW = Pingshuo water leached coal sample; PSM = Pingshuo HCl leached coal sample;

PSD = Pingshuo HF leached coal sample; SDR = Shendong raw coal sample; SDW = Shendong water leached coal sample;

SDM = Shendong HCl leached coal sample; SDD = Shendong HF leached coal sample

TABLE-2
ASH COMPOSITION ANALYSES OF COALS USED IN EXPERIMENT

Sample	Ash composition (wt%)									
	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃	TiO ₂	CaO	MgO	SO ₃	K ₂ O	Na ₂ O	P ₂ O ₅
PSR	48.3	44.7	2.3	2.0	1.3	0.1	0.2	0.2	0.1	0.5
SDR	23.3	9.7	16.0	0.8	27.5	3.2	14.2	0.1	2.3	0.0

TABLE-3
CATION CONCENTRATIONS IN DIFFERENT LEACHING SOLUTION

Sample	Process	Different cation forms (μg/L)					
		K ⁺	Na ⁺	Ca ²⁺	Mg ²⁺	Fe ³⁺	Al ³⁺
Pingshuo coal	Water-leaching	0.08	0.09	0.452	0.14	<0.0001	0.32
	HCl-leaching	2.86	1.54	16.42	2.16	4.56	62.30
	HF-leaching	0.43	0.35	6.11	0.28	12.89	238.11
Shendong coal	Water-leaching	0.19	1.21	1.712	1.95	<0.0001	0.32
	HCl-leaching	0.39	17.58	141.00	17.06	14.11	34.47
	HF-leaching	0.42	27.81	162.71	40.67	21.21	108.56

until no chloride ion was detected and dried at 80 °C in a vacuum oven for 24 h. For HF leaching process, the 10 g HCl-leached coal was mixed with 75 mL HF (40%). The slurry was stirred for 24 h at room temperature, filtered, leached with ultrapure water and dried at 80 °C in the vacuum oven for 24 h. The raw coal sample was denoted as R, the water-leached sample was denoted as W, the HCl-leached sample was denoted as M and the HF-leached sample was denoted as D.

The filtrate with different solvents from the fractional step washing coal process were metered volume to 500 mL of volumetric flask, the K⁺, Na⁺, Ca²⁺, Mg²⁺, Fe³⁺ and Al³⁺ concentrations were detected by atomic absorption spectrometer (Varian, U.S.). The results of the analysis were presented in Table-3.

Apparatus and procedure: An STA 409C TGA apparatus (Netzsch, Germany) was used. Approximately 20 mg sample was placed on the platinum pan, which was suspended from a thin quartz rod attached to one arm of the microbalance. The system was purged with high purity nitrogen prior to the heating of sample from room temperature to 1000 °C at a constant heating rate of 10 °C/min. The carrier gas was N₂, with a flow rate of 60 mL/min.

RESULTS AND DISCUSSION

Effects of pyrolysis characterization on the fractional step demineralization: The TG and DTG curves of Pingshuo

coal and Shendong coal raw coals and their demineralized samples during pyrolysis are illustrated in Fig. 1. It can be seen that the weight losses of the samples gradually increase with increasing temperature, but the change trend between raw coal and the samples leached by different solution for the same coal is different and there also exists difference between Pingshuo coal and Shendong coal. We can see easily that the weight losses of PSR and PSW are obviously higher than that of PSM and PSD, which indicates that the removing minerals by HCl leaching have the catalytic role on the pyrolysis for Pingshuo coal. Meanwhile, the TG curves of PSM and PSD are almost identical, which suggests that the retaining minerals after HCl leaching almost have no effect on the pyrolysis reactivity of Pingshuo coal. In order to investigate which mineral can be effectively removed by different leaching processes respectively and further review which mineral has the catalytic role on the pyrolysis, the AAS analysis was used to measure the metal cations in the leaching solution and the results can be seen in Table-3. The results show that the majority of alkali and alkaline earth metals (AAEM) minerals can be removed by the HCl leaching process, which indicates that these alkali and alkaline earth metals minerals play a catalytic role on the pyrolysis for Pingshuo coal. For Shendong coal, the weight loss of SDD is evidently lower than that of other three coals for Shendong coal. It reveals that the retaining

Fig. 1. TG and DTG curves of Pingshuo coal and Shendong coal during pyrolysis in N_2 atmosphere

minerals in SDM after HCl leaching have catalytic role on the pyrolysis of Shendong coal, which is different from Pingshuo coal. From Table-3, we can see that most of alkali and alkaline earth metals minerals can be removed by HF leaching process, in addition to Al-containing minerals. Compare with the Pingshuo coal, the most of Al-containing minerals can be removed by HF leaching process, but have no effect on the pyrolysis reactivity of Pingshuo coal. Therefore, we can infer that the Al-containing minerals have an inert role on the pyrolysis. In other words, only the alkali and alkaline earth metals minerals have a catalytic role on the pyrolysis of two coals, which is consistent with the published work^{3,4}. As mentioned above, Shendong coal, as the weakly reductive coal, has some specific properties compare with Pingshuo coal. Table-2 showed it shows that lower-level silicon and aluminum, higher-level alkali metals, alkaline earth metals and iron contents are found in SDR. Generally, minerals in coal exist in two forms of included and excluded forms. Some minerals including alkali metals or alkaline earth metals exist in coal in the included form, as the primary and secondary mineral. They closely link with the organic matrix in coal and are embedded in the coal fabric. Maybe those catalytic minerals cannot fully removed by HCl leaching process for Shendong coal.

Based on the DTG curves from Fig. 1, we can see that there is a similar weight loss peak in the 350-500 °C temperature

range, which is the primary devolatilization phase. These peaks are identical for the raw coal and the samples leached from Pingshuo coal and Shendong coal, respectively. From *ca.* 500 to 1000 °C, as the secondary devolatilization phase, the peak shapes and weight loss amounts vary with the pre-treatment of samples. PSR and PSW exhibit obvious weight losses at *ca.* 600 °C, but no observable peak is shown for PSM and PSD at this temperature range. For the Shendong coal, samples exhibit obvious weight loss peaks at *ca.* 600 °C, except for SDD. The weight loss between 200 and 350 °C, mainly from the volatilization of the mobile phase and the thermal decomposition of weak bonds, is small, which is in the range of 1.0-3.1 wt%. Weight loss mainly takes place in the temperature range between 350 and 500 °C, attributed to the primary devolatilization phase. In this phase, the weight loss of Pingshuo coals is within 12.3-13.9 wt%, while 10.6-11.7 wt% for Shendong coals. It is clear that the weight loss in this phase have no obvious difference for different leaching process samples from Pingshuo coal and Shendong coal. Thus, it indicates that those alkali and alkaline earth metals minerals have no effect on the pyrolysis activity in this phase. The proportion of weight loss between 500 and 1000 °C varies with the different leached samples from Pingshuo coal and Shendong coal. There is an evident change of weight loss for Pingshuo coal before and after HCl leaching process, the weight loss is decreased from 31.6 wt% for PSW to 16.0 wt% for PSM, it also indicates that those alkali and alkaline earth metals minerals removed by HCl leaching have catalytic role on the pyrolysis activity in this phase. For the Shendong coal, there is an obvious decrease from 27.7 wt% for SDM down to 19.6 wt% for SDD. This makes clearly that those alkali and alkaline earth metals minerals removed by HF leaching have catalytic role on the pyrolysis activity for Shendong coal, which is inconsistent with Pingshuo coal. For the water leaching process, the weight loss is increased after water leaching for two coals. Table-3 showed that there is only a few metal cations exist in the water leaching solution. A small amount of water-soluble carbonates, sulfates, chlorides, physically absorbed inorganics¹² or dust can be removed by water leaching process, which plays a role in cleaning the surface of coal. It should be a reason why the weight loss increases after water leaching for two coals.

Kinetics analysis of coal pyrolysis: The kinetic analysis has been studied to further reveal the effect of fractional step demineralization on the coal pyrolysis. It was assumed that coal pyrolysis is a first order reaction¹³ and then the kinetic parameters of coal pyrolysis were determined by the integral method⁴. So the coal pyrolysis reaction equation may simply be expressed as following formula:

$$\frac{dx}{dt} = A \exp\left(-\frac{E}{RT}\right)(1-x) \quad (1)$$

where A is pre-exponential factor, E is activation energy, T is temperature, t is time, x is pyrolysis conversion which can be calculated by:

$$x = \frac{(W_o - W_t)}{(W_o - W_f)} \quad (2)$$

where W_0 is the original mass of the test sample; W_t is the mass at time t and W_f is final mass at the end of pyrolysis.

For a constant heating rate H during pyrolysis, $H = dt/dt$, rearranging Eq. (1) and integrating gives

$$\ln \left[\frac{-\ln(1-x)}{T^2} \right] = \ln \left[\frac{AR}{HE} \left(1 - \frac{2RT}{E} \right) \right] - \frac{E}{RT} \quad (3)$$

Since it may be shown that for most values of E and for the temperature range of the pyrolysis, the expression $\ln[AR/HE(1-2RT/E)]$ in eqn. 3 is essentially constant, if the left side of eqn. 3 is plotted vs. $1/T$, a straight line may be obtained if the process can be assumed as a first-order reaction. Based on the slope and intercept of the straight line, the activation energy E and the pre-exponential factor A can also be determined⁴.

The plots of $\ln[-\ln(1-x)/T^2]$ versus $1/T$ is shown in Fig. 2, it can be seen that the reaction of coal pyrolysis should be described by three independent first order reaction at different temperature regions. The kinetic parameters and weight loss of coal samples at different stages is shown in Table-4, we can see that there are good correlation coefficient for all lines, which indicates that three independent first order reaction model fits the experimental data very well. From Table-4, it can be seen that the fractional step demineralized coals have different activation energy and pre-exponential factor with two raw coals. We can see that PSR and PSW have almost equivalent activation energy. After HCl leaching, the activation energy of PSM is obvious increased, that of PSD has almost unchanged compared with the PSM. For Shendong coal, the activation energy of fractional step demineralized coals is also higher than that of SDR. Normally, high activation energy means that the

Fig. 2. Plots of $\ln(-\ln(1-x)/T^2)$ vs. $1/T$ of PS and SD coals pyrolysis calculated by one-step integral method

reaction needs more energy from the surroundings. Thus, the increasing of activation energy of the fractional step demineralized coals also proves that those inherent minerals removed by the demineralization have the catalytic characteristics on the pyrolysis. It is consistent with the above analysis, the catalytic role of the alkali and alkaline earth metals minerals on the coal pyrolysis.

Conclusion

Based on the above analysis and discussion, the following conclusions can be drawn:

(1) Inherent minerals can be successful removed by the fractional step demineralization, most of alkali and alkaline earth metals minerals in Pingshuo coal were removed by the HCl leaching, only HF leaching can remove the Al-containing minerals, which is different from Shendong coal.

(2) The removed alkali and alkaline earth metals minerals in coal play a catalytic role on pyrolysis characteristics, which mainly present in the secondary devolatilization phase.

(3) The change of activation energy of the fractional step demineralized coals also proves the catalytic role of removed inherent minerals on the pyrolysis of two coals.

TABLE-4
KINETICS PARAMETERS OF PINGSHUO COAL
AND SHENDONG COAL PYROLYSIS

Sample	Temp. (°C)	Δm (wt%, daf)	E (KJ/mol)	A (min ⁻¹)	-R ^a
PSR	282-377	1.54	27.0	2.16	0.994
	377-469	9.82	63.4	2708.06	0.998
	469-899	25.91	13.3	4.18	0.965
PSW	283-371	1.17	31.6	5.23	0.989
	371-472	9.53	63.4	2657.19	0.997
	472-899	28.38	13.0	16.04	0.968
PSM	280-372	2.05	37.3	7.64	0.998
	372-473	10.53	70.4	5773.60	0.995
	473-898	17.15	17.4	1.22	0.975
PSD	278-375	2.40	40.9	12.29	0.997
	375-470	9.75	73.0	7542.64	0.997
	470-898	17.02	19.8	1.18	0.973
SDR	266-371	2.03	21.4	0.97	0.997
	371-469	8.27	36.1	14.24	0.999
	469-899	26.34	14.3	1.13	0.979
SDW	266-371	2.15	30.4	4.95	0.999
	371-469	5.04	44.3	79.19	0.997
	469-899	31.05	13.6	1.44	0.953
SDM	266-371	2.59	35.1	7.07	0.999
	371-472	7.65	50.1	151.06	0.999
	472-899	24.55	14.8	5.26	0.977
SDD	282-371	2.45	31.2	3.45	0.999
	371-472	8.24	48.3	106.26	0.999
	472-897	19.23	18.0	1.11	0.991

^a R, correlation coefficient.

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