# Study on the Effect of Coal Drying in Adsorbing $CO_2$ at Different Temperature and pH

by

Suhana bte Muhamad

A project dissertation submitted to the Chemical Engineering Programme Universiti Teknologi PETRONAS in partial fulfillment of the requirement for the BACHELOR OF ENGINEERING (Hons) (CHEMICAL ENGINEERING)

JANUARY 2005

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# CERTIFICATION OF APPROVAL

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# CERTIFICATION OF ORIGINALITY

This is to certify that I am responsible for the work submitted in this project, that the original work is my own except as specified in the references and acknowledgements, and that the original work contained herein have not been undertaken or done by unspecified sources or persons.

Where Mahared

MŚ. SUHANA BTE MUHAMAD

#### ABSTRACT

CO<sub>2</sub> is the primary green house gas representing roughly 83% of the anthropogenic effect. One of the options to mitigate the rising atmospheric concentration of CO<sub>2</sub> is through CO<sub>2</sub> geologic sequestration in coal seams process. Part of understanding the sequestration is to study what is the effect on the rate of CO<sub>2</sub> adsorption at different temperature and pH in a coal mine environment. The objectives of this study are to investigate CO<sub>2</sub> gas adsorption patterns on local coal sample at different temperature, pH and particle sizes. It is also to determine the basic properties of the local coal sample and determine how the characteristic of the coal sample affects the adsorption rate of CO<sub>2</sub> gas. In the characterization of the coal sample process, there were a number of parameters which was studied and tested. They were the moisture content, the ash content, carbon content and mineral content of the coal sample. For the study on the effect of varying parameters on the CO<sub>2</sub> adsorption on coal seams, the CO<sub>2</sub> adsorption behavior was investigated using a manometric apparatus. The experimental set-up will be used in the investigation of the effects of temperature (24.6°C, 30°C, 40°C and 55°C) pH (acidic of pH 0.51, near neutral of pH 5.97 and alkali of pH 12.40), and particle size (1000µm and 2000µm) of the CO<sub>2</sub> adsorption rate on the local coal sample. The moisture content of the coal sample was found to be 37.4%. The ash content analysis gave a percentage ash of 11.02%. The elemental composition analysis gave an elemental carbon content of 56% to 60%. The mineral matter in the coal sample was found to be 12.09%. The chemical elements analysis indicated silica as having the highest amount in the coal sample. The experiments conducted for the study of the CO<sub>2</sub> adsorption in coal seams showed that smaller particle size 1000µm had a higher adsorption rate per mass of coal sample as compared to the 2000µm sample. The effect of increasing the temperature is to decrease the equilibrium adsorption capacity of the coal samples. It can be observed that the untreated coal sample has the highest extent of adsorption capacity followed by the acidic, alkali and neutral conditions. The coal samples were found to be of lignite type. From literature review, coals of lignite type were found to exhibit the most sorption tendency towards CO<sub>2</sub> as compared to coals of other ranks.

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# CHAPTER 1 INTRODUCTION

#### **1.1 BACKGROUND OF STUDY**

Carbon dioxide ( $CO_2$ ), which makes up for the majority of the green house gas, was seen to have increased in its concentration in the atmosphere over the years. Our main concern here is the global warming effect which will cause worldwide problems such as ozone reduction and desertification.

In our efforts to control the  $CO_2$  concentration rise, researches have been conducted in many areas. Among them are the sequestrations of  $CO_2$ . This generally means taking control over the concentration of  $CO_2$  in the atmosphere. Basically there are a few types of sequestration. They are ocean sequestration, geologic sequestration and terrestrial sequestration. Ocean sequestration is where  $CO_2$  are stored in oceans through direct injection or fertilization. Geologic sequestration is carried out by using natural pore spaces in geologic formations which serve as reservoirs for long term carbon dioxide storage while the third type of sequestration, terrestrial sequestration, is where a large amount of carbon is stored in soils and vegetations, which are our natural carbon sinks. Among these three types of sequestration, geologic sequestration is thought to have the largest potential for near-term application.

Geologic sequestration of  $CO_2$  has also been recognized as an environmentally attractive method to reduce the amount greenhouse gas emissions. Of the geologic options, sequestering  $CO_2$  in coal seams has several advantages. Coal, which is formed through the physical and chemical alteration of peat (coalification) by processes involving bacterial decay, compaction, heat and time, is an agglomeration of many different complex hydrocarbon compounds. Composed of carbon, hydrogen, oxygen, nitrogen, sulfur and other elements, coal can be divided in four major ranks, anthracite coal, bituminous coal, sub-bituminous coal and lignite coal. Coal seams, which are actually the cracks and fractures on the surface of coal, contain natural gas hydrocarbon compound, formed as a by-product of the coal formation process. These gases, for example methane gases, are also known as coal seam gas (CSG). The basic idea of  $CO_2$  sequestration in coal seams utilizes the basic idea of replacing these coal seam gases with  $CO_2$  gases in the atmosphere.

Among the advantages of  $CO_2$  sequestration in coal seams are that  $CO_2$  injection can enhance methane production from coal beds, coal can trap  $CO_2$  for long periods of time, and potential major coal basins that contain ideal beds for sequestration are near many emitting sources of  $CO_2$ . All these compensations contribute to the factor why coal seams option of sequestering  $CO_2$  gas is so appealing.

### **1.2 PROBLEM STATEMENT**

# **1.2.1 Problem Identification**

 $CO_2$  is the primary green house gas representing roughly 83% of the anthropogenic effect. One of the options to mitigate the rising atmospheric concentration of  $CO_2$  is through  $CO_2$  geologic sequestration in coal seams process. In the research of performing geologic sequestration in unminable coal mine, part of understanding the sequestration is to study what is the effect on the rate of  $CO_2$  adsorption at different temperature and pH in a coal mine environment. The main purpose of this research is to study the adsorption patterns of  $CO_2$  gas in coal seams in an environment created in the laboratory that best represent the environment in a coal mine where the actual sequestration will take place. The main two parameters that will be investigated are on the variation of temperature and pH. It is also crucial that the characteristics of the coal are studied upon since the rate of adsorption of  $CO_2$  will differ for each different type of coal.

## 1.2.2 Significant of Project

The sequestration of the carbon dioxide gas  $(CO_2)$  in coal seams is seen as a possible way to mitigate the rising atmospheric concentrations of  $CO_2$ . Technologies that have been developed for enhanced oil recovery and enhanced coal bed methane recovery could be applied to the long term disposal of  $CO_2$ . In order to determine which coal seams would be good to act as the disposal sites and under what environmental conditions the sequestered  $CO_2$  would remain stable, a better understanding is needed of the chemistry of the coal- $CO_2$  system.

This research will study the effects of temperature and pH variation on the adsorption rate of  $CO_2$  gas in coal seams which will provide us crucial information of the optimum environment needed for effective adsorption process of  $CO_2$  to occur in coal mines.

### **1.3 OBJECTIVES AND SCOPE OF STUDY**

### 1.3.1 Objectives

The objectives of this study are:

- 1. To investigate CO<sub>2</sub> gas adsorption patterns on local coal sample at different temperature.
- 2. To investigate  $CO_2$  gas adsorption patterns on local coal sample at different pH.
- 3. To investigate  $CO_2$  gas adsorption patterns on local coal of different particle sizes.
- 4. To determine the basic properties of the local coal sample and determine how the characteristics of the coal sample affects the adsorption rate of CO<sub>2</sub> gas.

#### 1.3.2 Scope of Study

The scope of study, as outlined by the objectives above, includes investigating the  $CO_2$  gas adsorption patterns on local coal under temperature variation. The temperatures tested were at 24.6°C, 30°C, 40°C and 55°C.

For the investigation of the  $CO_2$  gas adsorption patterns on local coal at different pH, conditions of acidic, near neutral and alkaline with the pH values of 0.51, 5.97 and 12.40 were created for the experiment. This experiment monitors the trend of the adsorption rate of  $CO_2$  in the local coal sample under different pH conditions.

The effects which the coal particle sizes have on the adsorption rate of  $CO_2$  gas were studied upon. Two sizes of coal particles were experimented on (1000µm and 2000µm in diameter).

The general properties of the local coal sample were also studied upon for the purpose of identifying and predicting the performance of  $CO_2$  adsorption in it. The properties investigated were the moisture content, ash content and carbon content.

Similarly, the chemical element analysis, which gives us the approximate amount of chemical content of the coal sample, can then be compared with other coal samples' chemical elemental analysis obtained from literature and give us a clearer understanding of how the adsorption rate is affected by this factor. The chemical elements analysis was conducted using the Atomic Absorption Spectrometer (AAS).

### 1.3.3 The Relevancy of Project

The extent to which coal can adsorb  $CO_2$  is affected by a number of factors. The nature of the coal will determine the maximum adsorption capacity under a given set of conditions, but the sequestration environment will determine the extent to which that ultimate capacity will be realized. The effects of both physical and chemical changes need to be understood. Parameters such as temperature, pressure and pH is expected to have a moderate to large influence, therefore creating the need for it to be studied upon (K. Schroeder et al).

The investigations on the variation of temperature and pH on the adsorption rate of  $CO_2$  on the coal sample will enable us to predict the optimum temperature and pH conditions for the best adsorption rate of  $CO_2$  gas in the geologic sequestration process to occur. Once the optimum conditions are determined, efforts can be made to identify the unminable coal mines whose conditions fit the identified optimum conditions. This location identification is a crucial first step in ensuring the success of the sequestration process.

In the absence of external influences, underground temperatures tend to be constant over time but increases with depth. The adsorption of  $CO_2$  is exothermic and will provide a heat source, at least during the active pumping phase of sequestration. Thus, it is important to know how temperature will affect the  $CO_2$  adsorption onto coal and whether the magnitude of this effect is universal for all coals or is rank or maceral dependent (K. Schroeder et al). The effects of temperature on the adsorption rate on the different types of coals can be compared through information obtained through literature. From here, we can compare the local coal sample with those from around the world and establish a prediction of the performance of the local sequestration process, should the project be implemented here in Malaysia.

Even if initially dry, coal seams will become wet as a result of drilling operations, fracturing of the coal bed and over lying strata, and the deposition of a combustion gas which may contain residual water of combustion. Thus, an aqueous phase will be present and will vary in composition according to its source and the nature of the coal bed and the surrounding material with which it is in contact. In natural systems, pH is often an important parameter and will change during sequestration. Little is known about the potential effect of a pH change on the ability of the organic matrix to adsorb CO<sub>2</sub>. It is well recognized that adsorption of CO<sub>2</sub> on solid surfaces is affected by the pH of the surrounding media (K. Schroeder et al). The extent to which pH changes will affect the CO<sub>2</sub> adsorption capacity of coals has not been investigated thoroughly, therefore need to be established to allow us to predict and take the necessary steps in ensuring the smooth operation of CO<sub>2</sub> sequestration process.

Depending on the capture technology, the  $CO_2$  stream may be as nearly pure as the  $CO_2$  a raw combustion gas, or something in between. In addition, gases such as hydrogen, methane, ethane and higher hydrocarbons may be present in the coal seam and act to inhibit or enhance the  $CO_2$  sequestration. These gases may displace  $CO_2$  and thus limit the durability of the sequestration (K. Schroeder et al). Therefore, it is imperative that the preferential sorption phenomena in coal seams between  $CO_2$  and  $CH_4$  gas are investigated. The  $CH_4$  gas is chosen here due to the fact that it makes up for the majority

of the gases underground and thus established itself as the main opponent of the  $CO_2$  for the adsorption process in coal seams.

Coal contains a wide variety of organic and mineral phases in a complex, porous, 3dimensional network which varies from one coal deposit to another and from one location to another within the same seam. The organic portion of the coal is thought to capture  $CO_2$  via surface adsorption, pore filling and solid solution. Less recognized is the possibility that the mineral phases present in the coal may assist via mineral carbonate formation. Thus, the nature of the coal seam itself is an important variable to be considered (K. Schroeder et al). This is where the chemical elements analysis comes into the picture. By analyzing the chemical elements of the local coal sample, a general definition of the predicted performance of the  $CO_2$  sequestration in the local coal seams can then be established.

Knowledge of the extent to which coals can adsorb  $CO_2$  under a variety of conditions is necessary to evaluate the long-term storage capacity of candidate seams. The nature of the coal will determine its maximum adsorption capacity, but the dynamic nature of the sequestration environment will determine the extent to which that capacity can be realized. In order to evaluate the long-term storage capacity of a coal seam, possible changes in the sequestration environment need to be anticipated and their effect understood (K. Schroeder et al). The experiments will allow us to study the  $CO_2$ adsorption rate under many possible environments scenarios in a coal mine. The data obtained from the experiments will enable us to have a better analysis and understanding of the  $CO_2$  sequestration process which will, in time, pave the way towards actually implementing such project in Malaysia.

# CHAPTER 2 LITERATURE REVIEW AND THEORY

### 2.1 COAL PROPERTIES

Coal is a heterogenous sedimentary rock, with physical and chemical properties determined by biochemical and geochemical transformation of vegetable matter. The extent of this transformation, known as coalification, is given by the rank of the coal, which is determined on the basis of a chemical (proximate) analysis. The proximate analysis involves the determination of the weight percent of moisture, volatile matter, fixed carbon and ash in the coal under prescribed heating. Coal ranking is a measure of the degree to which the original vegetable matter approaches the structure of pure carbon conditions (M.E.272 Standford University). Rank is a classification of coal beds that indicates the degree of metamorphism, or progressive alteration, from lignite to anthracite (R. Stanton et al).

Anthracite coal is of the highest metamorphic rank. It is also known as "hard" coal and has a brilliant luster. It can be rubbed without leaving a "familiar coal dust" mark on the finger and can even be polished for use as jewelry. Anthracite coal burns slowly with pale blue flame and may be used primarily as a domestic fuel. Bituminous coal burns with a smoky flame and may also contain 15-20% w/w volatile matter. It is the most abundant variety of coal, weathers only slightly, and can be kept in open piles with very little danger of spontaneous combustion. Sub-bituminous coal is not as high on the metamorphic scale as bituminous coal and has often been called "black lignite". Lignite is the coal that is lowest on the metamorphic scale. It may vary in color from brownblack and is often distinguished from the sub-bituminous coals as having lower carbon content and a higher moisture content. Lignite may dry out and crumble in air and is certainly liable to spontaneous combustion (Speight, 1994, p.5).

Material	Wood %	Peat %	Lignite %	Sub- Bituminous Coal %	Bituminous Coal %	Anthracite %
Moisture	30-60	90+	20-40	10-20	13-1	2-3.5
(material as						
found)						
Moisture (air-	10-15	20-25	15-25	10-20	13-1	2-3.5
dried material)						
Dry, ash-free						
material:						
Carbon	50	55-65	65-73	73-78	78-92	92-96
Hydrogen	6.0	5.5	4.5	6.0	5.3	2.5
Oxygen	43	32	21	16	8	4

Table 2.1: Analyses in the Coalification Series

Table is obtained from Combustion and Gasification of Coal, A Williams; M. Pourkashanian; J. M Jones; N. Skorupska; pg.28.

As coal rank increases from lignite to anthracite, the density of the coal initially decreases from lignite to high volatile bituminous coal rank as a result of expulsion of water and compaction and the formation of micropores. In low rank coals, which are less than 75 percent C, on a dry, ash-free (daf) basis (lignite and sub-bituminous ranks), surface area has been interpreted as primarily contained in macropores > 20 nm (R. Stanton et al).

In contrast, density of the high volatile bituminous to anthracite coals increases as a result of coalification processes that drive off hydrogen and oxygen. Pores in these higher rank coals are primarily micropores (<2nm) and to a lesser extent transitional pores (2-20nm) (R. Stanton et al).



Figure 2.1: Density Trend in Coal Ranks

Because the adsorption capacity is both a function of the amount and reactivity of surface area contained in pores and possibly fractures, understanding the relationships of adsorption of  $CO_2$  and coal rank as well as with the coal composition is of crucial importance. Rank is known to have an effect on the amount of carbon dioxide that can be adsorbed into the coal porosity (R. Stanton et al).

Structurally, coal can be viewed as being composed of water, minerals and organic components (macerals). Organic material dominates, typically representing 85 - 95% (wt/wt) of a dry coal. These organic materials occur in various different petrographic types, called "maceral" which reflect the nature of the precursor plant material. Various inorganic materials comprise 5 - 15% of the coal. A third structural element, and perhaps the most distinctive feature when compared to other solid fossil fuel sources like petroleum and oil shale, is an extensive network of pores. These pores give high surface area (>100 m<sup>2</sup>/g for bituminous and sub-bituminous coals and lignites) and an appreciable volume of pore space, allowing access to a significant fraction of the organic material (Levine et al, 1981).



Figure 2.2: Diagram of the major constituents in coal: organic material, fragments of plant debris (macerals), inorganic inclusions, and an extensive pore network.

Macerals are the most combustible components of coal and can vary in chemical composition both among maceral varieties within a single coal bed and among coal beds. The general groups of macerals are vitrinite, liptinite and inertinite. Macerals originate from partially decomposed plant parts that are altered through the coalification process. Their dominant chemical composition is primarily a mixture of carbon, hydrogen, oxygen, nitrogen and sulfur. Of the major maceral groups, liptinite is richer in hydrogen while inertinite is richer in carbon than vitrinite (R. Stanton et al).

As a result of differing peat-forming conditions and plant assemblages, organic components are transformed into macerals which arrange structurally into lithotypes that, in turn, comprise facies or the major subunits of a coal bed. Adsorption properties of coal beds vary with the composition of the coal. Gases should be adsorbed most by vitrain-rich facies that contain low amounts of minerals. Physical properties of a coal bed can also be of significant factor in the adsorption of  $CO_2$ . The properties include fracture intensity, degree of fracture filling, overburden and seat rock permeability, coal rank, reservoir pressure and the degree of folding and faulting of the coal bed (R. Stanton et al).

The most permeable facies are those rich in fusain layers and the most impermeable layers contain discrete mineral layers, impure coal facies that are rich in minerals, or attrital lithotypes (R. Stanton et al).



Figure 2.3: Structural relationships among macerals and lithotypes that comprise facies of a coal bed.

The percentage of water in coal (inherent moisture) is used to differentiate coals by rank up to the high volatile bituminous stage. Low rank coals contain more water than higher rank coals. Water content is significant in coals from low to medium rank, ranging from  $\approx 40\%$  (wt/wt) for typical lignite to  $\approx 14\%$  (wt/wt) for bituminous coal (Levine et al, 1981).

Among the minerals found in coals are Phosphate Pentoxide ( $P_2O_3$ ), Silica (SiO<sub>2</sub>), Ferric Oxide (Fe<sub>2</sub>O<sub>3</sub>), Alumina (Al<sub>2</sub>O<sub>3</sub>), Tilania (TiO<sub>2</sub>), Lime (CaO), Magnesia (MgO), Sulfur Trioxide (SO<sub>3</sub>), Potassium Oxide (K<sub>2</sub>O) and Sodium Oxide (Na<sub>2</sub>O) (M.E 272 Stanford University). These are among the major minerals that make up the structure of coal. The general representative percentage and the mineral constituents are given below. The major elements that make up for these minerals are also included here.

Constituents	Representative Percentage
SiO <sub>2</sub>	40 - 90
Al <sub>2</sub> O <sub>3</sub>	20-60
Fe <sub>2</sub> O <sub>3</sub>	5-25
CaO	1 – 15
MgO	0.5 - 4
Na <sub>2</sub> O	0.5 - 3
K <sub>2</sub> O	0.5 - 3
$SO_3$	0.5 - 10
P <sub>2</sub> O <sub>5</sub>	0-1
TiO <sub>2</sub>	0 - 2

Table 2.2: Major Inorganic Constituents of Coal Ash

Table is obtained from The Chemistry and Technology of Coal, Second Edition, James G. Speight.

Constituent	Range
Aluminum	0.43 -3.04%
Calcium	0.05 - 2.67%
Iron	0.34 - 4.32%
Potassium	0.02 - 0.43%
Magnesium	0.01 - 0.25%
Sodium	0.00 - 0.20%
Silicon	0.58-6.09%
Titanium	0.02 - 0.15%

Table 2.3: Some of the Major Elements in Coals

Table is obtained from The Chemistry and Technology of Coal, Second Edition, James G. Speight.

A wide variety of pore sizes exists in coal of different ranks, ranging from large  $(\geq 300 \overset{\circ}{A})$  macropores to small  $(\leq 12 \overset{\circ}{A})$  micropores. Knowledge of the total volume of

such pores, the pore size distribution, and the effective surface area of the organic material is needed to anticipate coal's potential reactivity (Levine et al, 1981).

### 2.2 EXPERIMENTAL METHODOLOGY

From literature reviews, there were some experimental methodologies found for the purpose of studying the adsorption of  $CO_2$  in coal seams. These methods has been reviewed and modified to suit with the available apparatus in the laboratory for the purpose of the study.

(A. Busch, Y. Gensterblum, B.M. Krooss) on their study of Methane and  $CO_2$  Sorption and Desorption Measurements on Dry Argonne Premium Coals: Pure Components and Mixtures, had utilized the volumetric gas adsorption method. The experimental set-up consisted of a stainless steel reference and sample cell, actuator-driven valves and a high precision pressure transducer. The cells are placed in a thermostated water bath for temperature calibration. The coal samples are kept in a stainless steel sample cell with a calibrated volume. The experiments are carried out by first evacuating both the sample and reference cell, before a certain amount of gas are admitted to the reference volume and calibrated. From the experiments, the pressure reductions readings in the sample cell containing the coal sample are taken. Using an equation of state (EOS), the amount of substance (moles of gas) in the reference cell is computed from the pressure, temperature and volume of the cell. The reduction in the number of moles indicates the amount of  $CO_2$  gas adsorbed in the coal sample.

(K. Schroeder, E. Ozdemir, B.I Morsi ) in their study of Sequestration of Carbon Dioxide in Coal Seams had also used the manometric principle in their experimental set-up which consisted of a reference and sample cell of known volume which was placed in a water bath. The sample cell was pressurized from the reference cell. Using the change in pressure in the reference cell and accounting for the gas compressibility, the number of moles of gas transferred from the reference cell was calculated. The number of gas-phase moles in the sample cell after the gas transfer was calculated from the post-transfer sample-cell pressure in a similar manner. The missing moles of gas were accounted to the adsorption of  $CO_2$  onto the coal.

# 2.3 EFFECT OF PARAMETERS VARIATION ON CO2 ADSORPTION

# 2.3.1 Effect of Temperature

From the experiments conducted under the research of K. Schroeder, E. Ozdemir and B.I Morsi, it was found that increased temperature caused a decrease in the equilibrium adsorption capacity of the coal. The tested temperatures were of  $22^{\circ}$ C,  $30^{\circ}$ C,  $40^{\circ}$ C and  $55^{\circ}$ C.

It was also found that the incremental amount of  $CO_2$  that can be adsorbed by the coal drops off at higher pressures. From a practical stand-point, this meant that a higher pumping cost per pound of  $CO_2$  will be incurred at higher sequestration pressures. In the pressure-temperature region studied (0 – 700 psi and 0 – 55<sup>o</sup>C), the CO<sub>2</sub> capacity did not exceed 1.4mmole  $CO_2/g$  coal. This adsorption value correspond to a  $CO_2$  storage density of about 85kg  $CO_2/m_3$  coal.

### 2.3.2 Effect of pH

(K. Schroeder, E. Ozdemir and B.I Morsi) had also included pH variation effects in their study of  $CO_2$  sequestratio since the pH of an aqueous solution affects the surface of a material it is in contact with.

The major difference observed from the experiments was the lower extent of adsorption for the samples that were treated, irrespective of the treatments. Among the treated samples, it was found that both acid and base treatment resulted to an increase in adsorption capacity compared to the coal treated with only de-ionized water.

# 2.3.3 Effect of Mineral Content in Coal

From the pH variation experiments conducted by K. Schroeder, E. Ozdemir and B.I Morsi, the acid treated sample was found to be lower in its mineral content due to the removal of acid-soluble minerals during the treatment process. The increased in the adsorption of  $CO_2$  for the acid treated sample was identified to be due to the increased accessible pore volume which was caused by the removal of mineral content from the coal sample. This finding establishes a relationship between the adsorption capabilities of a coal sample with its mineral content where lower mineral content containing coals will be able to adsorb  $CO_2$  better as compared to coals with high mineral content.

# CHAPTER 3 PROJECT METHODOLOGY

## **3.1 COAL CHARACTERIZATION IDENTIFICATION**

## **3.1.1 Experimental Procedure Utilized**

The project involves identifying the general nature and type of the coal sample obtained from the coal mines in Sarawak. Since there was no information given regarding the characteristics of the coal, this characterization identification was a crucial step in determining and predicting the effects on the  $CO_2$  adsorption as critical parameters (e.g. temperature, pH) are varied. This is due to the fact that coals of different rank and mineral content will behave differently to the adsorption process. In other words, this coal characterization step will enable us to predict the performance of the  $CO_2$ sequestration if it were to be implemented on our local coals here in Malaysia.

In the characterization process, there were a number of parameters which was studied and tested. They were the moisture content, the ash content, carbon content and mineral content of the coal sample. The procedures for each of these experiments are discussed below.

### **3.1.1.1 Moisture Content Analysis**

The coal sample (passing 250 $\mu$ m sieve) was taken and placed in a pre-weighed Petri dish and weighed. The Petri dish containing the coal sample was then placed for an hour inside an oven at a temperature of 110<sup>o</sup>C. The sample was then weighed again to determine the weight loss from the heating process. The loss of weight from the coal sample was taken as the moisture content of the coal sample.

#### 3.1.1.2 Ash Content Analysis

The coal sample was grinded to pass a  $150\mu m$  sieve. With an open 50mL quartz/high silica crucible used as the container, approximately 6g of this grinded coal sample was placed in a cold muffle furnace and heated gradually till the temperature reaches  $300^{\circ}C$  in

an hour. The heating process was continued till the temperature reached  $500^{\circ}$ C in the second hour. Ashing was continued for another 2 hours at  $500^{\circ}$ C. The sample was stirred once each hour. At the end of the 2 hours, the sample was cooled and reweighed, before being grinded and reignited at  $500^{\circ}$ C for an additional hour. The resulting sample was cooled and reweighed again to calculate the percentage ash.

# 3.1.1.3 Elemental Composition Analysis (Carbon, Hydrogen, Nitrogen, Sulfur and Oxygen Content Analysis)

The CHNS-O analysis was performed on the coal sample using the CHNS-O Equipment. 5 samples were tested which consisted of the coal sample with the particle size of  $<150\mu m$ . The coal sample of  $<150\mu m$  was oven dried first to remove the moisture from the sample prior to the experiment.

### **3.1.1.4 Chemical Element Analysis**

The coal sample was analyzed for its chemical element content. From literature review, it was known that coals will generally contain some major minerals which are Phosphate Pentoxide ( $P_2O_3$ ), Silica (SiO<sub>2</sub>), Ferric Oxide (Fe<sub>2</sub>O<sub>3</sub>), Alumina (Al<sub>2</sub>O<sub>3</sub>), Tilania (TiO<sub>2</sub>), Lime (CaO), Magnesia (MgO), Sulfur Trioxide (SO<sub>3</sub>), Potassium Oxide (K<sub>2</sub>O) and Sodium Oxide (Na<sub>2</sub>O). Through the use of the Atomic Absorption Spectrometer (AAS), these chemical elements were detected.

The usage of the AAS required the coal sample to be in liquid form. Thus, the coal sample had to be liquefied first using a method obtained from the American Standard of Testing Materials (ASTM) 2000. The preparation of reagents needed was done prior to the dissolution process. The following reagents were prepared as follows:

- a) Aqua Regia Solution One part of concentrated nitric acid (HNO<sub>3</sub>) was mixed with three parts of concentrated hydrochloric acid (HCl) and one part water.
- b) Boric Acid Solution, Saturated 60g of boric acid (H<sub>3</sub>BO<sub>3</sub>) was dissolved in 1L of de-ionized water.

- c) Stock Solutions, Standard Standard stock solutions were prepared from high-purity metals, oxides, or salts.
- d) Hydrofluoric Acid Concentrated hydrofluoric acid (HF).

The dissolution process was carried out by placing approximately 0.2 g of ashed coal sample in a 25mL sample containers. 3mL of Aqua Regia solution and 5mL of HF was added to this. With the screw caps tightened, the samples were then heated at a temperature of  $130^{\circ}$ C for at least 2 hours after which the resultant solution was then transferred to 100mL cylindrical flasks and 50mL of saturated H<sub>3</sub>BO<sub>3</sub> solution was added. The solution was cooled to room temperature and their volume was adjusted to 100mL by the addition of de-ionized water. Blanks were also prepared in a similar manner. This final solution obtained was then tested against standard solutions prepared for the identified metal components using the AAS.

## **3.2 CO<sub>2</sub> ADSORPTION EXPERIMENTS**

### 3.2.1 Experimental Set-up

For the study on the effect of varying parameters on the  $CO_2$  adsorption on coal seams, the following experimental apparatus was constructed.



Figure 3.1: Experimental Set-up of Equipment.

The  $CO_2$  adsorption behavior was investigated using an almost similar manometric technique as stated in the literature review. Basically, 2 gas bombs are connected together by plastic tubes with a ball valve. The ball valve functions to control the gas flow from the first reference bomb to the second bomb (containing coal sample). The reference bomb here was used primarily for the purpose of controlling the pressure in the system. At the other end of the sample bomb, another ball valve is placed. This valve functions to either prevent the gas from escaping to the atmosphere and to provide an outlet connection of the experimental apparatus. At the end of each bomb, a digital manometer is placed for the purpose of pressure measurement and monitoring. The experimental set-up is placed in a temperature controlled water bath.

The above experimental set-up will be used in the investigation of the effects of temperature, pH and particle size of the  $CO_2$  adsorption rate on the local coal sample.

### **3.2.2 Particle Size Variation Experiment**

For the particle size variation, the coal sample was grinded and sieved into different particle sizes ranging from  $< 150\mu$ m to  $2000\mu$ m. The particle size effect on the adsorption rate was conducted using the particle sizes of  $1000\mu$ m and  $2000\mu$ m at a temperature of  $40^{0}$ C. The smaller sized particles ( $<150\mu$ m) was utilized for other experimental purposes involving the determination of the sample coal characteristics.

### **3.2.3 Temperature Variation Experiment**

For the temperature variation, the temperatures of 24.6  $^{0}$ C, 30  $^{0}$ C, 40  $^{0}$ C and 55  $^{0}$ C were tested. The temperatures were set so as to give the best representation of the reservoir conditions. The coal sample used here was of the 1000 $\mu$ m.

# 3.2.4 pH Variation Experiment

For the pH variation, the effect of the coal surface pH was investigated by pre-soaking the coal using aqueous slurries at pH values of approximately 0.51, neutral, 12.54 and untreated (acidic, neutral, alkali and untreated conditions respectively). The coals were then filtered and dried overnight in an oven at  $60^{\circ}$ C before being experimented on. The

experiment was conducted at temperature of  $40^{0}$ C, so as to simulate the reservoir conditions. The coal sample used here was of the 1000 $\mu$ m.

It should be noted that the coal samples were evacuated in an oven at 80<sup>o</sup>C overnight prior and after conducting the experiments.

#### **3.2.5 Experimental Procedure**

The coal samples to be tested using the experimental set-up were first treated by evacuating them in an oven overnight at  $80^{\circ}$ C. This is done so as to remove any entrapped gases in the coal seams and to allow the maximum adsorption of CO<sub>2</sub> to occur.

The coal samples were placed in a metal bomb container (approximately more than  $\frac{3}{4}$  full) and after tightening the bomb, it was then connected to the rest of the apparatus. Note here that the water bath was set to the desired water temperature prior to the loading of the coal containing sample cell. Extra care was taken here to ensure that the connections were leakage proof. Once the experimental set-up was completed, nitrogen gas (N<sub>2</sub>) was flowed through the system. This was done for approximately 3 minutes to purge out any gases in the system prior to the adsorption process.

Once purged,  $CO_2$  gas was then flowed through the system. The flow rate of  $CO_2$  and the opening of the outlet valve (Valve 2) were adjusted so as to maintain a stable pressure inside the system. The stable pressure was achieved once valve 2 can be closed completely without any connections in the system popping open. This stable pressure was usually achieved approximately at 60kPa.

With the stable pressure inside the system, the temperature calibration of the  $CO_2$  gas was then done by closing off the connecting valve between the reference and sample cell (Valve 1). This closure allows the containment of  $CO_2$  gas inside the sample cell system and also calibrates its temperature. The pressure in the sample cell system was then observed. Once it has reached approximately 17kPa, the pressure readings were taken once every minute till the pressure reaches 0kPa.

Once the pressure readings were taken, the sample cell was then disconnected from the apparatus, and the weighing of the coal sample was done. The coal sample were then placed in an oven and evacuated overnight at  $80^{\circ}$ C.

The pressure readings taken were corrected to 17kPa for the purpose of synchronizing the readings. The pressure reductions for each of the readings were then calculated and divided by the mass of the coal sample. This gave the pressure reduction of CO<sub>2</sub> per mass of coal sample, which was then plotted against time for the purpose of studying the trend of the adsorption process.

# CHAPTER 4 RESULTS AND DISCUSSIONS

## 4.1 COAL CHARACTERIZATION IDENTIFICATION - RESULTS

The experiments conducted for the purpose of the sample coal characterization were the moisture, ash, carbon, and mineral content analysis. These experiments were conducted so as to determine the type and characteristics of local coal sample. This determination is of crucial importance since coals of different rank, for example, has different  $CO_2$  adsorption trends.

## 4.1.1 Moisture Content Analysis

The following results were obtained from the moisture content analysis conducted on the local coal sample.

	1 <sup>st</sup>	2 <sup>nd</sup>	3 <sup>rd</sup>	A	
	Reading	Reading	Reading	Average	
Weight of Petri Dish - empty	65.0570	65.0575	65.0571	65.0572	
(g)					
Weight of Petri Dish + Coal	92,6499	92,6490	92.6484	92.6491	
Sample (g)	<b>JI</b> (0) (J) J				
Weight of Petri Dish + Coal	82 3304	82,3381	82 3433	82,3373	
Sample after heating (g)	02.0001	02.0001			
Weight of Water (g)	92.6491-82.3373=10.3118				
Percentage of Water		10.3118	$\times 100\% - 37$	10/	
(Moisture Content) (%)	$\frac{1}{92.6491 - 65.0572} \times 100\% = 37.4\%$				

Table 4.1: Results from Moisture Content Analysis

# 4.1.2 Ash Content Analysis

The following are the results of the ash content analysis of the local coal sample.

	1 <sup>st</sup>	2 <sup>nd</sup>	3 <sup>rd</sup>	
	Reading	Reading	Reading	Average
Weight of high-silica crucible	45,2935	45 2933	45 2935	45 2934
$(x_{i})$ (g)	10.22555	10,4755	10.2955	13.2934
Weight of coal sample $(x_2)$ (g)		6.	0048	L
Weight of crucible + coal				
sample after first cooling (x <sub>3</sub> )	45.6451	45.6466	45.6472	45.6463
(g)				
Weight of crucible + coal				
sample after second cooling $(x_4)$	45.6458	45.6463	45.6469	45.6463
(g)				
Percentage ash (%)	$= \frac{\left( \left[ x_1 + x_2 \right] - \left[ \frac{x_3 + x_4}{2} \right] \right)}{\left( x_1 + x_2 \right)} \times 100\%$ $= \frac{51.2982g - 45.6463g}{51.2982g} \times 100\% = 11.02\%$			

Table 4.2: Rest	ults from Ash	Content Analysis
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# 4.1.3 Elemental Composition Analysis (Carbon, Hydrogen, Nitrogen, Sulfur and Oxygen Content [CHNS-O])

The following are the results from the CHNS-O analysis conducted.

	Carbon	Hydrogen	Nitrogen	Sulfur	Oxygen
	(%)	(%)	(%)	(%)	(%)
Run 1	56.31	4.071	1.900	0.364	26.335

Table 4.3: Results from CHNS-O Analysis

Run 2	58.46	4.285	1.957	0.402	23.876
Run 3	57.74	4.330	1.922	0.379	24.609
Run 4	56.17	3.909	1.888	0.300	26.713
Run 5	60.24	4.200	1.889	0.279	22.372

Note: The oxygen percentage was calculated by subtracting the sum of the other components of ultimate analysis from 100 [% Oxygen = 100 - (%C + %H + %N + %S + %Ash] (ASTM D3176).

### 4.1.4 Mineral Matter Percentage

Using the results obtained from the ash percentage and the average percentage of sulfur compound from the elemental compound analysis, the percentage of the mineral matter can be calculated as below using the Parr formula. Here, [A] is the ash percentage while [S] is the sulfur percentage taken as the average:

% Mineral matter = 
$$1.08A + 0.55S = 1.08(11.02) + 0.55(0.3448) = 12.09\%$$
.

### 4.1.5 Chemical Elements Analysis

The following results were obtained from the Atomic Absorption Spectrometer (AAS) for the purpose of detecting the metal content in the coal sample. These metals make up for the minerals in the coal sample. The results obtained from the analysis are shown below.

Metal	Concentration (ppm)	<b>Concentration (%)</b>
Calcium (Ca)	3.14	$3.14 \times 10^{-4}$
Sodium (Na)	2.95	$2.95 \times 10^{-4}$
Magnesium (Mg)	1.31	1.31×10 <sup>-4</sup>
Potassium (K)	12.40	$1.24 \times 10^{-3}$
Iron (Fe)	23.94	2.394×10 <sup>-3</sup>
Silica (Si)	436.34	4.363×10 <sup>-2</sup>
Aluminum (Al)	155.9	$1.559 \times 10^{-2}$

Table 4.4: Results from the Atomic Absorption Spectrometer (AAS) Analysis

# 4.2 COAL CHARACTERIZATION IDENTIFICATION - DISCUSSIONS

The summary of the obtained results from the above analysis are given below.

Analysis	Results
Moisture Content	37.4 %
Ash Content	11.02 %
Mineral Matter	12.09 %
	Carbon ~ 56 – 60 %
	Hydrogen ~ 3.9 – 4.3 %
Elemental Composition	Nitrogen ~ 1.89 – 1.95 %
	Sulfur $\sim 0.28-0.4$ %
	Oxygen ~ 22.4 – 26.3 %

Table 4.5: Summary of Results from Analysis of Coal Sample

### 4.2.1 Coal Rank

In general sense, coal can be defined as an organic rocklike natural product with its resemblance to rock is due to its physical nature and composition. The kinds of plants from which the coal originated, the kinds of minerals inclusions, and the nature of the maturation conditions that prevailed during the metamorphosis of the plant material gives rise to different coal types. The rank of a coal refers to the degree of metamorphosis. For example, coal that has undergone the most extensive change or metamorphosis has the highest rank (determined from the fixed carbon or heating value). The grade of a coal refers to the amount and kind of inorganic material (mineral matter) within the coal matrix (Speight, J.G).

Coal contains significant proportions of carbon, hydrogen and oxygen with lesser amounts of nitrogen and sulfur. Attempts have been made to classify coal on the basis of elemental composition and one of the early classification systems based on the elemental composition of coal (done by Seyler) was subsequently extended (Francis, 1961). For coals below the anthracite rank and with an oxygen content less than 15%, it was possible to derive a relationship between carbon content (C, %w/w), hydrogen content (H, %w/w), caloric value (Q, cal/g), and volatile matter (VM, %w/w). However, their applications are limited due to applicability to only specific coal types. Nonetheless, this classification does offer an initial attempt to quantify coal behavior (Francis, 1961).

Coal rank is often equated directly with carbon content, since rank does progress from high-carbon coal to low-carbon coal (or vice-versa). However, other properties of the coal are also taken into consideration in assigning rank. The American Society for Testing and Materials (ASTM, 1991) has evolved a method of coal classification that relies on the fixed carbon content of the coal as well as other physical properties. This table is included in the appendix of the report.

This system of classification indicates the degree of coalification as determined by methods of proximate analysis, with lignite being classed as low-rank coal and anthracite as high-rank coal. Thus, coal rank increases with the amount of fixed carbon (which is the solid residue, other than mineral ash, remaining after all of the volatile matter has been removed under prescribed conditions and calculated by subtracting the percentages of moisture, volatile matter and ash from 100%) but decreases with the amount of moisture and volatile matter. This is why coal rank is often equated to the proportion of elemental carbon. Although it is true that anthracite coal usually contains more carbon than bituminous coal, which in turn contains more carbon than sub-bituminous coals are not so well defined as for the fixed carbon, and extreme precaution was advised in attempting to equate coal rank with proportion of elemental carbon (Speight, J.G).

From the results of the experiments conducted, it was found that the local coal sample contained a very high value of total moisture content (37.4%). From literature review, it was generally stated that high values of moisture content in coals indicates a lower ranking coal type, with lignite having a value of approximately 20 - 40%.

The elemental composition analysis gave an approximation range of 56.31% to 60.24% of the elemental carbon content in the coal sample. This, together with the moisture content analysis enabled the determination of the rank of the coal sample. From the literature review done, these results indicate that the coal sample are of lignite type, which is the lowest rank in the coal ranking system. Although precautions were advised in the literature reference by J.G Speight in equating coal rank with proportions of elemental carbon instead of the fixed carbon value, this approximation was seen to be fit since the moisture content of the coal sample were found to very high and the oxygen content of the coal sample falls within the 20% - 30% range, which is the lignite type range. These facts therefore support the conclusion that the coal samples studied were of lignite type.

### 4.2.2 Mineral Matter

Mineral matter in coal originates from the inorganic constituents of the vegetation that acted as the precursor to the coal and from the mineral matter that was transported to the coal bed from any remote site. Thus, mineral matter in coal has often been classified as inherent or extraneous (Francis, 1961; Stach et al., 1982). Inherent mineral matter is defined as mineral matter that had its origin in the organic constituents of the plants giving rise to the coal bed, whereas extraneous mineral matter is matter that was brought into the coal-forming deposit by mechanical means from outside.

Mineral matters in coals are usually determined indirectly, with the ash analysis forming the basis of the calculation. Ash is the residue derived from the mineral matter during complete incineration of the coal. However, determination and chemical analysis of the ash content gives the average content of the inorganic elements in it. Nevertheless, ash analysis can provide valuable data that, when used with data from other sources, may give a representation of the mineral content of the coal (Speight, J.G). Thus, the Parr formula was utilized to assess the mineral matter in coal. The mineral matter content of coal varies considerably and may even be as high as 35% of the coal by weight.
Through the ash content analysis, it was found that the coal sample had a percentage ash of 11.02%. From the ash content and the average sulfur content obtained from the elemental composition analysis, the percentage of mineral matter in the coal sample was determined and found to be approximately 12.09%.

The evaluation of coal mineral matter was taken further by determining the individual metal constituents of the ash. This was done by utilizing the Atomic Absorption Spectrometer (AAS) in which spectra are obtained for standards and samples aspirated into a flame. The obtained results are shown below. It can be seen here that Silica (Si) exists in the largest amount in the coal sample, followed by Aluminum (Al), Iron (Fe), Potassium (K), Calcium (Ca), Sodium (Na) and finally Magnesium (Mg). However, these elements are present in very small amounts if they were compared to the generally found amounts of elements given in the literature review. Nonetheless, the order of which they exist (quantitatively) follows quite closely the general major inorganic constituents of coal ash obtained from literature review. Here, it can be concluded that the coal sample was found to contain small amount of the tested elements. The calculated mineral matter of 12.09% may consist of other elements, which could explain why the tested elements were found to be small in its quantity.

Analysis	Results
	Calcium ~ 3.14ppm
	Sodium ~ 2.95ppm
	Magnesium ~ 1.31ppm
Chemical Element	Potassium ~ 12.40ppm
	Iron ~ 23.94ppm
	Silica ~ 436.34ppm
	Aluminum ~ 155.9ppm

Table 4.6: Results of Chemical Elements Analysis

#### 4.2.3 CO<sub>2</sub> Adsorption in Coals as a Function of Rank and Composition

From literature review, based on a research done by Robert C. Burruss under the USGS research on geologic sequestration, it was found that lower ranking coals have a larger ratio of  $CO_2/CH_4$  adsorption as compared to higher ranking coals. The lower ranking coals have a greater tendency to adsorb  $CO_2$  compared to  $CH_4$ . This meant that lower ranking coals are better candidates for the purpose of  $CO_2$  sequestration compared to higher ranking coals. It was also found that increase in the moisture content of the coal increases the  $CO_2/CH_4$  adsorption ratio. This meant that higher moisture containing coal will perform better in adsorbing  $CO_2$ . There was no obvious correlation found for the  $CO_2$  adsorption with rank and composition. However, it was observed that the  $CO_2$  adsorption is highest for anthracite coals followed by lignite, bituminous and subbituminous.

Discussing the results obtained with the literature review above, it can be concluded that the local coal sample is a very good candidate to be utilized as the medium for  $CO_2$  sequestration. This is based on the fact that the coal was identified as lignite type and has high moisture content which, from the literature review, is the mark of a good coal type to adsorb  $CO_2$ .

## **4.3 CO<sub>2</sub> ADSORPTION EXPERIMENTS**

### **4.3.1 Introduction**

The main purpose of conducting the  $CO_2$  adsorption experiments was to investigate how the rate of the adsorption process is affected when certain identified critical parameters which are the particle size, temperature and pH conditions of the studied coal sample are varied.

Generally speaking, adsorption processes is where one or more components of a gas or liquid stream are adsorbed on the surface of a solid adsorbent and a separation is accomplished (Geankoplis, C. J.). It involves transfer and resulting equilibrium distribution of one or more solutes between fluid phase and particles. The portioning of a single solute between fluid and sorbed phases or the selectivity of a sorbent towards multiple solutes makes it possible to separate solutes from a bulk fluid phase or from one another. Adsorption involves, in general, the accumulation (or depletion) of solute molecules at an interface. The accumulation per unit surface area is small, thus, highly porous solids with very large internal area per unit volume are preferred. Adsorbent surfaces are often physically and/or chemically heterogenous, and bonding energies may vary widely from one site to another.

The adsorption separation is based on three distinct mechanisms; steric, equilibrium and kinetic mechanisms. In the steric separation mechanism, the porous solid has pores having dimension such that it allows small molecules to enter while excluding large molecules from entry. The equilibrium mechanism is based on the solid having different abilities to accommodate different species, that is the stronger adsorbing species is preferentially removed by the solid. The kinetic mechanism is based on the different rates of diffusion of different species into the pore, thus by controlling the time of exposure the faster diffusing species is preferentially removed by the solid (Do, Duong D.).

There are basically two types of adsorption. The first being physisorption, which involves van der Waals forces. This type of adsorption occurs when the intermolecular forces between molecules of a solid and the gas are greater than those between the gas molecules itself. The second type of adsorption is of the chemisorption which occurs through the formation of chemical bonds between the gas and coal.

In our case, the type of adsorption occurring are of the physisorption type. The intermolecular forces between the  $CO_2$  gas (adsorbate) and the coal, which acts as the adsorbent, are stronger as compared to the intermolecular forces between the  $CO_2$  molecules themselves. This promoted the adsorption of the  $CO_2$  into the coal seams and hence created the opportunity for us to study the possibility of utilizing this simple process for the purpose of  $CO_2$  sequestration to mitigate the rising  $CO_2$  concentration in the atmosphere.

## 4.3.2 Experimental Procedure Justification

The experimental procedure utilized the manometric principles which are often utilized in  $CO_2$  sequestration in coal seams researches all over the world. The main reason behind this common utilization is due to the fact that the study applies the philosophy of the nonideal gas law to measure the number of molecules adsorbed by the coal sample. This simple principle is shown below.

$$PV = ZnRT$$
[1]

Where:

P = Pressure of Sample Cell (kPa)
V = Volume of Sample Cell (m<sup>3</sup>)
n = Number of Moles of Gas in Sample Cell (moles)
R = Universal Gas Constant (8.314 kJ/kmol.K)
T = Temperature of the Sample Cell (<sup>0</sup>C)
Z = Compressibility factor for Sample Gas

Rearranging the equation will result in the following:

$$P = \left[\frac{ZRT}{V}\right]n$$
 [2]

Assuming the  $\left[\frac{ZRT}{V}\right]$  factor remains constant, the pressure of the sample cell P, can be observed to be directly proportional to the number of CO<sub>2</sub> gas moles in the sample cell n. Thus, relating to the conducted study, the pressure reduction readings taken during the experiment, when plotted against the time period, were actually directly proportional to the loss in the number of CO<sub>2</sub> gas in the system. This is shown below.

$$\Delta P = \left[\frac{ZRT}{V}\right] \Delta n \qquad [3]$$

Hence, the trends observed in the plotted graphs can be taken as representatives of the rate of adsorption of  $CO_2$  gas into the coal sample seams, which is our main focus in the research and thus justifies the method used in the experiments.

## 4.3.3 Experimental Operating Conditions Justification

The operating conditions of the conducted experiments (particle sizes, temperature and pH conditions) were decided upon based on literature review done on similar researches conducted and also upon the ease and practicality of the experimental set-up.

The temperature and pressure chosen for the experiments were of fundamental importance since both play a crucial part in distinguishing the physical properties of the  $CO_2$  gas. Thus, proper care was taken in choosing the operating temperature and pressure so as to maintain the  $CO_2$  gas in gas phase. This was done by choosing and maintaining both of these parameters below the critical point of  $CO_2$  gas (31°C and 1070 psia). Increase in the pressure of the gas at temperatures below the critical point will result in condensation of the gas and cause in an apparent increase in the adsorbed amount of  $CO_2$  gas in the system. This will clearly result in inaccurate data and thus was avoided.

Another factor considered when choosing these parameters are the practicality and feasibility of the operating parameters with the experimental set-up. Since the set-up had to be constructed independently by the author, it could not handle high pressure. The maximum pressure attainable in the system was around 80kPa. Thus, the conducted experiments were conducted below this maximum system pressure.

The choices for temperatures the purpose of studying the variation effects were based upon the decision to simulate the reservoir temperatures as closely as possible. From literature reviews done on the subject of  $CO_2$  sequestration, the temperature was set at the maximum of  $55^{\circ}C$  to simulate the reservoir temperature. This indicates that the temperature is higher in the coal mines. However, there were also some researches which conducted the experiments at lower temperatures of  $22^{\circ}C$ . Thus, the temperatures were chosen to be within these two values with the lowest temperature  $24.6^{\circ}C$ , due to limitations of the water bath temperature. For the particle size and pH condition variations, the temperature of  $40^{0}$ C and coal particle sizes of  $1000\mu$ m was utilized for the sole purpose of synchronizing the experiments.

The experiments conducted for the study of the  $CO_2$  adsorption in coal seams consisted of varying the parameters of particle size, temperature and pH of the coal samples. The results of these experiments are discussed below.

## 4.3.4 Particle Size Variation Experiment

The results for the particle size variation experiments are plotted below.



Effects on CO2 Adsorption in Coal Seams at Different Particle Sizes

Figure 4.1: Effect on CO<sub>2</sub> Adsorption in Coal Seams at Different Particle Sizes (1000µm and 2000µm) at 17kPa.

The effect of different particle sizes on the  $CO_2$  adsorption in coal seams were investigated using coal samples of two different particle sizes, 1000µm and 2000µm. As shown in the plotted graph in Figure 4.1, the smaller particle size 1000µm had a higher adsorption rate per mass of coal sample as compared to the 2000µm sample.

As in any other subject matter, smaller particles size will result in a larger surface area of the material. In this case, a larger surface area caused the coal seams to be more exposed to the  $CO_2$  gas, thereby enhancing the rate of adsorption. This can be observed by the higher rate of pressure reduction per mass coal sample as plotted.

However, for the purpose of  $CO_2$  sequestration, it should be noted that although smaller particles of a given mass of coal sample will have a greater surface area as compared to larger particles (thus enhancing the rate of adsorption), the coal's huge internal surface area dominates so much that the particle size essentially has no effect on the capacity of the coal to adsorb  $CO_2$ . In other words, the particle size of the coal will have not much of a significant impact on the rate of  $CO_2$  adsorption in large scale operations.

Comparing the results obtained from the conducted experiments on the identified lignite coal sample with the data of previously conducted research done on bituminous coal sample, the following graph was obtained. Here, the Indonesian coal is of bituminous type while the local coal sample is of lignite type.



Comparison of Adsorption Rate for the Same Particle Size Between Local and Indonesian Coal Sample

Figure 4.2:  $CO_2$  Adsorption for local lignite and Indonesian bituminous (both of the same particle size of 1000 $\mu$ m and temperature of 40<sup>o</sup>C) at approximately 15kPa.

As observed above, the local coal sample, being a lower ranking coal sample, as compared to the Indonesian bituminous coal sample, gives a higher adsorption rate, hence confirming the literature review on the  $CO_2$  adsorption in coals as a function of rank and composition discussed earlier. The same trend was observed for both of the temperatures and pH conditions experiments as well. This was also proven to be true in many experiments conducted by researchers for the purpose of  $CO_2$  sequestration. This means that the prospects of implementing the  $CO_2$  sequestration project in Malaysia is greater since the coal sample can adsorb  $CO_2$  better.

#### **4.3.5 Temperature Variation Experiment**

The results for the temperature variation experiments are plotted and discussed below.



#### Effects on CO2 Adsorption in Coal Seams at Different Temperatures

Figure 4.3: Effect on CO<sub>2</sub> Adsorption at Different Temperatures (24.6 $^{\circ}$ C, 30 $^{\circ}$ C, 40 $^{\circ}$ C and 55 $^{\circ}$ C) at 17kPa.

The temperature variation experiments were conducted at different temperatures of 24.6  $^{0}$ C, 30  $^{0}$ C, 40  $^{0}$ C and 55  $^{0}$ C. The plotted graph for these experiments is shown in Figure

4.3. From the graph, it can be observed that the effect of increasing the temperature is to decrease the equilibrium adsorption capacity of the coal samples.

For the sake of showing that the adsorption process reaches its equilibrium state (starts to level off), another set of experiments were conducted using higher pressure. The temperatures tested here are of  $30^{\circ}$ C and  $40^{\circ}$ C. The trend of the adsorption rate can also be seen here to be decreasing with higher temperatures. The data obtained is plotted below.



Figure 4.4: Effect on  $CO_2$  Adsorption at Different Temperatures (30<sup>o</sup>C and 40<sup>o</sup>C) – Reaching Equilibrium at 60kPa.

This decrease can be explained using Langmuir, who was the first to propose a coherent theory of adsorption onto a flat surface based on a kinetic viewpoint, that there is a continual process of bombardment of molecules onto the surface and a corresponding evaporation (desorption) of molecules from the surface to maintain zero rate of accumulation at the surface at equilibrium. The assumptions of the Langmuir model are that the surface is homogenous, that is the adsorption energy is constant over all sites (Do, Duong D.). It assumes that the adsorption on surface is localized, that is adsorbed atoms or molecules are adsorbed at definite, localized sites and each site can accommodate only one molecule or atom. The Langmuir theory is based on a kinetic principle, that is the rate of adsorption (which is the striking rate at the surface multiplied by a sticking coefficient, sometimes called the accommodation coefficient) is equal to the rate of desorption from the surface.

Equating the rates of adsorption and desorption, the following Langmuir isotherm was obtained.

$$\theta = \frac{bP}{1+bP}$$

Where;

$$b = b_{\infty} \exp\left(\frac{Q}{R_g T}\right)$$

Here, Q is the heat of adsorption and is equal to the activation energy for desorption,  $E_d$ . The parameter b is called the affinity constant or Langmuir constant and is a measure of how strong an adsorbate molecule is attracted onto a surface.



Figure 4.5: Behavior of the Langmuir Equation (Adsorption Analysis: Equilibria and Kinetics, Duong D. Do)

When the affinity constant b is larger, the surface is covered more with adsorbate molecules as a result of stronger affinity of adsorbate molecule towards the surface. Similarly, when the heat of adsorption Q increases, the adsorbed amount increases due to the higher energy barrier that adsorbed molecules have to overcome to evaporate back to the gas phase. Increase in the temperature will decrease the amount adsorbed at a given pressure. This is due to the greater energy acquired by the adsorbed molecule to evaporate.

This trend in the decreasing equilibrium adsorption capacity of the coal samples can also be explained from the Le Chatelier's principle. The change in temperature causes changes in the equilibrium constant K. A temperature rise will increase K for an endothermic process and decrease K for an exothermic reaction. Discussing this in our case point of view, the adsorption process of  $CO_2$  onto the coal sample is an exothermic process which can be indicated by the following. Here, the endothermic reaction is also included for the purpose of discussions.

> Reactants  $\leftrightarrow$  products + heat (Exothermic reaction) Reactants + heat  $\leftrightarrow$  products (Endothermic reaction)

If we consider heat as a component in the equilibrium system, a rise in temperature "adds" heat to the system and a drop in temperature "removes" heat from the system. As with a change in any other component, the system shifts to reduce the effect of the change. Therefore, a temperature increase favors the endothermic (heat-absorbing) direction and a temperature decrease favors the exothermic (heat-releasing) direction. In our case, since the adsorption is an exothermic process, the production of heat will cause an imbalance in the equilibrium of the system. To attain back this equilibrium, the system will proceed in a different direction to achieve stability. Hence, with increasing temperatures, the heat in the system becomes much larger, causing the reaction to proceed more and more to the left side of the equation to attain equilibrium. This resulted in less adsorption to occur which explains the decrease in the equilibrium adsorption capacity observed from the graph.

Putting these reasoning into perspective, this means that otherwise equivalent, deeper, warmer seams will adsorb less  $CO_2$  at a given pressure than shallower, cooler ones (K. Schroeder et al).

## 4.3.6 pH Variation Experiment

The results for the pH variation experiments are plotted and discussed below.



#### Effect of CO2 Adsorption in Coal Seams at Different pH Conditions

Figure 4.6: Effect on CO<sub>2</sub> Adsorption at Different pH Conditions (Acidic, Neutral, Alkali and Untreated) at 17kPa.

The variation of pH conditions experiments were conducted on coal samples that were treated with acid solution, distilled water (neutral), alkaline solution, and untreated sample. It can be observed that the untreated coal sample has the highest extent of adsorption capacity followed by the acidic, alkali and near neutral conditions.

From literature review, at pH of 9 (alkali), dissolved  $CO_2$  exists as the carbonate ion and mineral carbonates are stable. At a pH of 2 (acidic) mineral carbonates dissolve, the  $CO_2$ that can remain dissolved exists as  $H_2CO_3$  and  $CO_{2(aq)}$  and  $CO_{2(gas)}$  is produced. The pH of an aqueous solution also affects the surface of a material it is in contact with. In the case of coal, at higher pH values, the carbonaceous surface, being in an environment above its isoelectric point, assumes a net negative charge. At lower pH value, the carbonaceous surface assumes a net positive charge. While a pH 9 will favor, and a pH 2 will disfavor, the aqueous capture of  $CO_2$ , the effect of being above, or below, the carbonaceous isoelectric point is not so obvious (K. Schroeder et. al).

The results obtained indicate lower extent of adsorption for those samples that were treated, regardless of treatment. Because all of these samples were oven dried whereas the untreated sample was not, it may not be valid to state that such mild drying would not affect the results. This is because moisture removal was known to affect the porosity and transport properties of coals, a phenomenon most commonly observed in lignite and subbituminous coals. However, the effects are usually mild for higher rank coals when the drying temperature is below  $100^{0}$ C (K. Schroeder et. al). Because of the uncertain effect of the drying, it is reasonable to limit the discussion of the pH effects to only three of the treated coal samples.

Comparing the rate of adsorption of  $CO_2$  gas between the pH conditions of acidic, alkaline and neutral, it can be seen that the acidic conditions has the highest adsorption capacity. From literature, coal samples that were treated with acid were found to have reduced in its acid soluble minerals such as calcite. It was also found that the sample showed an ash reduction. If the removal of acid-soluble minerals resulted in increased accessible pore volume, then it is expected that the  $CO_2$  capacity would increase. The ash content of base treated samples was found to have reduced by a smaller percentage. In this case, it seems more likely that any increase in capacity would be due to surface adsorbed NaOH. If water is also present, alkali metal hydroxides react with  $CO_2$  to form carbonates. Whatever the mechanism, acid treatment appears to have a greater effect than the base or neutral treatments in these experiments (K. Schroeder et. al).

It should be noted here that from literature review, for experiments conducted using wet samples, the alkali treated samples showed a slightly higher adsorption rate as compared to the acid treated one. However, for similar experimental conditions, the opposite result was obtained if dry samples were used. This proves that the effect of the coal samples being above or below its carbonaceous isoelectric point is not so obvious. In other words, the rate of adsorption for acid and alkali treated samples will be more or less the same.

Nonetheless, whatever condition the samples were in (either wet or dry), the similarities found was that the untreated sample gave the highest rate while the neutral sample gave the lowest rate of adsorption. This phenomenon could be attributed to the fact that the untreated samples still retains its original moisture content which may have assisted the adsorption process. The high rate could also be attributed by the mineral phases present inside the coal through the process of mineral carbonate formation. As for the higher rate of adsorption of the acidic and alkali treated samples as compared to the neutral sample, this could be due to the fact that some of the acid/alkali soluble minerals present inside the coal had dissolved during the treatment process. This thus increased the accessible pore volume, and hence increases the adsorption rate slightly as compared to the neutral sample.

## CHAPTER 5 CONCLUSIONS AND RECOMMENDATIONS

## 5.1 CONCLUSIONS

The research on the effect of coal drying in adsorbing  $CO_2$  at different temperature and pH was conducted on the local coal sample obtained from the coal mines of Sarawak. The research had also indulged in the process of determining the coal rank as well as other general properties of the coal.

In a nutshell, the research had covered the determination the coal sample characteristics through the moisture, ash, elemental compositions, mineral matter and the chemical elements analysis. The moisture content was found to be 37.4% while the ash percentage was 11.02%. The elemental composition of the sample coal was found to consist of carbon (56 – 60 %), hydrogen (3.9 – 4.3 %), nitrogen (1.89 – 1.95 %), sulfur (.28 – 0.4 %) and oxygen (22.4 – 26.3 %). The results of chemical elements analysis indicates the following compositions of the elements tested, calcium (3.14ppm), sodium (2.95ppm), magnesium (1.31ppm), potassium (12.40ppm), iron (23.94ppm), silica (436.34ppm) and aluminum (155.9ppm). The total mineral matter was found to be 12.09%.

Through these analyses, the coal samples were found to be of lignite type. It is known that rank has an effect on the amount of carbon dioxide that can be adsorbed in the coal seams. From literature review, coals of lignite type were found to be among those that have the highest tendency towards adsorbing  $CO_2$  as compared to coals of other ranks.

For the experiments conducted for the purpose of studying the  $CO_2$  gas adsorption patterns on local coal sample at different temperature, pH and particle sizes, the following results are obtained. For the temperature variation (24.6  $^{\circ}C$ , 30  $^{\circ}C$ , 40  $^{\circ}C$  and 55  $^{\circ}C$ ), the rate of  $CO_2$  gas adsorption was found to have decreased with increasing temperature. This is due to the entropy term in the free energy expression. In the variation of the pH conditions, it was observed that the untreated coal sample exhibit the highest  $CO_2$  gas adsorption as compared to the three treated coal samples (acidic, alkaline and near neutral each having a pH value of 0.51, 5.97 and 12.40 respectively). Between these three treated coal samples, acidic treated samples were found to be higher in its  $CO_2$  gas adsorption followed by alkali and finally neutral. The high adsorption rate for the untreated sample may be due to the natural moisture contained inside the coal and also due to the mineral assistance via the mineral carbonate formation. The higher rate of adsorption for the acid and alkali treated sample as compared to the neutral sample could be due to the fact that some of the acid/alkali soluble minerals inside the coal had dissolved during the treatment process. This increased the accessible pore volume and hence increases its adsorption was higher for smaller coal particles (1000µm) as compared to larger particles (2000µm) due to larger surface area owned by the smaller particles. However, due to the coal's huge dominant internal surface area, this factor will have much of a significant effect in large scale projects.

Based on these findings, some general operating conditions for  $CO_2$  sequestration can be predicted and tabulated. For example, efforts should be given to maintain the sequestration location's temperature at the desired optimum point. This is crucial since the adsorption process itself is known to be exothermic and it was clearly indicated from the experimental results that higher temperatures will cause unfavorable conditions for  $CO_2$  adsorption to occur which in turn will lead to the incompletely filled coal seams. Similarly, the estimated costs of alternative sequestration scenarios, such as the sequestration of acid gases, need to take into account changes in the sequestration chemistry, such as increased pore capacity due to mineral dissolution or other surface changes.

### **5.2 RECOMMENDATIONS**

These recommendations are made for the purpose of improving the current project for future researches. Several improvements can and should be made in terms of planning and carrying out the experiments so that better overall outcome of the project can be achieved.

Future researchers into this topic should familiarize themselves with the coal sample itself. The basic procedures for determining the type of coal and its other properties should be known prior to the start of the research. This is to ensure that the researcher is well aware of the parameters and the expected results that he/she is searching for when conducting the experiments.

It is highly recommended that the test or experiments conducted on the coal sample be done in accordance with the standard procedures given in the American Standard for Testing Materials (ASTM) so as to ensure that accurate analysis of the coal sample can be obtained. The utilization of this standard will also allow comparison be done on a standard basis between the results obtained with the ones from literature review.

It is also recommended that the experimental set-up be improved so as to enable better adsorption process to take place and be monitored. The utilized experimental set-up was problematic in the sense that the coal sample had to be taken out of the sample cell and weighed separately after each experiment to determine its mass. This created some discrepancies in the mass value since some unavoidable loss of the coal sample will occur during the transfer, hence creating opportunities for errors to occur. A new experimental set-up should be utilized so as not to repeat the errors in future researches.

For future researches, it is highly recommended that the desorption rate of  $CH_4$  be studied upon at different parameter variations. Since the main attraction of  $CO_2$  sequestration in coal seams was because it enhances the  $CH_4$  production in coal beds, it is only logical that this matter was thoroughly investigated. It is also recommended that the investigation of the role of minerals in assisting the  $CO_2$  adsorption on coal seams be done. This can be carried out by testing the adsorption rate of coal samples of the same rank from different sources. The mineral content of the coal samples should be first determined and any relationship between the mineral content and the rate of adsorption can then be studied upon. Since the exact role of minerals has not yet been determined, it should be interesting to see if there are any significant correlations or connections that can be establish between these two.

Although from an economic point of view, the project for the purpose of  $CO_2$  sequestration solely may not be feasible for a small country like Malaysia, at the very least, the researches conducted can be utilized to enhance the production of  $CH_4$  gases from the coal mines. Through this, we will be able to kill two birds with one stone. Not only will we be able to increase the production of  $CH_4$ , which are economically feasible, we will also be doing our part in helping to reduce the  $CO_2$  concentration in the atmosphere and in doing so, keep the global warming effects at bay.

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## **APPENDICES**

## APPENDIX A: CO<sub>2</sub> ADSORPTION EXPERIMENTAL DATA

TABLE A1: Experimental Data for the Particle Size Variation on CO<sub>2</sub> Adsorption – Particle Size 1000µm.

T = 40°C Mass of coal sample: 95g

	-								_						
	Reduction of Pressure per Mass Coal (kPa/g)	0	0.019473684	0.040631579	0.059684211	0.076842105	0.092315789	0.106315789	0.119052632	0.130947368	0.141578947	0.151578947	0.160947368	0.170105263	0.178421053
	Reduction of Pressure(kPa)	0	1.85	3.86	5.67	7.3	8.77	10.1	11.31	12.44	13.45	14.4	15.29	16.16	16.95
	Corrected Data (kPa)	17	15.15	13.14	11.33	9.7	8.23	6.9	5.69	4.56	3.55	2.6	1.71	0.84	0.05
	Pressure (kPa)	17.19	15.34	13.33	11.52	9.89	8.42	7.09	5.88	4.75	3.74	2.79	1.9	1.03	0.24
Sample coal size: 1000µm	Time (min)	0	1	2	3	4	5	6	7	8	6	10	11	12	13

TABLE A2: Experimental Data for the Particle Size Variation on CO<sub>2</sub> Adsorption – Particle Size 2000µm.

T = 40°C Mass of coal sample: 95g Sample coal size: 2000.....

Sample coal size: zuuuli	II IIIIIIIIIIIIIIIIIIIIIIIIIIIIIIIIIII			
Time (min)	Dracellira (kDa)	Corrected Data (kDa)	Reduction of Presente(kDa)	Reduction of Pressure
0	17.08	2011 2012 2014 (m m) 17	0	
-	15.55	15.47	1.53	0.016105263
2	14.12	14.04	2.96	0.031157895
m	12.77	12.69	4.31	0.045368421
4	11.52	11.44	5.56	0.058526316
ۍ ا	10.33	10.25	6.75	0.071052632
9	9.14	9.06	7.94	0.083578947
7	8.15	8.07	8.93	0.094
ω	7.12	7.04	9:96	0.104842105
G	6.15	6.07	10.93	0.115052632
10	5.24	5.16	11.84	0.124631579
11	4.34	4.26	12.74	0.134105263
12	3.5	3.42	13.58	0.142947368
13	2.7	2.62	14.38	0.151368421
41	1.94	1.86	15.14	0.159368421
15	1.2	1.12	15.88	0.167157895
16	0.51	0.43	16.57	0.174421053

TABLE A3: Experimental Data for the Temperature Variation on CO<sub>2</sub> Adsorption – Temperature at 55<sup>0</sup>C

T = 55°C Mass of coal sample: 70.05g Sample coal size: 1000µm

Time (min)	Pressure (kPa)	Corrected Data (kPa)	Pressure(kPa)	per Mass Coal (kPa/g)
0	17.76	17	0	0
-	16.48	15.72	1.28	0.018272662
2	15.29	14.53	2.47	0.035260528
3	14.17	13.41	3.59	0.051249108
4	13.12	12.36	4.64	0.066238401
5	12.14	11.38	5.62	0.080228408
6	11.18	10.42	6.58	0.093932905
7	10.26	9.5	7.5	0.107066381
8	9.41	8.65	8.35	0.119200571
o	8.59	7.83	9.17	0.130906495
10	7.79	7.03	9.97	0.142326909
11	7.05	6.29	10.71	0.152890792
12	6.35	5.59	11.41	0.162883655
13	5.67	4.91	12.09	0.172591006
14	5.02	4.26	12.74	0.181870093
15	4.38	3.62	13.38	0.191006424
16	3.76	3	14	0.199857245
17	3.16	2.4	14.6	0.2084225555
18	2.59	1.83	15.17	0.2165596
19	2.04	1.28	15.72	0.224411135
20	1.51	0.75	16.25	0.231977159
21	0.96	0.2	16.8	0.239828694

TABLE A4: Experimental Data for the Temperature Variation on CO<sub>2</sub> Adsorption – Temperature at 40<sup>o</sup>C

# $T = 40^{\circ} \text{C}$

Mass of coal sample: 68.69g Sample coal size: 1000µm

đ	(																	
Reduction of Pressurv	per Mass Coal (kPa/g	0	0.022128403	0.042655408	0.062163343	0.080506624	0.097394089	0.113262484	0.128548551	0.143106711	0.1566458	0.169456981	0.181831417	0.193914689	0.205561217	0.216479837	0.22681613	
Reduction of	Pressure(kPa)	0	1.52	2.93	4.27	5.53	6.69	7.78	8.83	9.83	10.76	11.64	12.49	13.32	14.12	14.87	15.58	
	Corrected Data (kPa)	17	15.48	14.07	12.73	11.47	10.31	9.22	8.17	7.17	6.24	5.36	4.51	3.68	2.88	2.13	1.42	
	Pressure (kPa)	17.87	16.35	14.94	13.6	12.34	11.18	10.09	9.04	8.04	7.11	6.23	5.38	4.55	3.75	ę	2.29	
	Time (min)	0		2	m	4	ວ	Q	7	ω	σ	10		12	13	14	15	

TABLE A5: Experimental Data for the Temperature Variation on CO<sub>2</sub> Adsorption – Temperature at 30<sup>0</sup>C

 $T = 30^{0} C$ 

Mass of coal sample: 67.77g

Sample coal size: 1000µm

Time (min)	Pressure (kPa)	Corrected Data (kPa)	Reduction of Pressure(kPa)	Reduction of Pressure per Mass Coal (kPa/g)
0	17.42	17	0	0
-	13.94	13.52	3.48	0.051350155
2	10.98	10.56	6.44	0.095027298
3	8.41	7.99	9.01	0.132949683
4	6.16	5.74	11.26	0.166150214
5	4.14	3.72	13.28	0.195956913
6	2.31	1.89	15.11	0.222960012
7	0.64	0.22	16.78	0.247602184

TABLE A6: Experimental Data for the Temperature Variation on CO<sub>2</sub> Adsorption – Temperature at 24.5<sup>0</sup>C

 $T = 24.6^{\circ}C$ 

Mass of coal sample: 67.63g Sample coal size: 1000um

	g) (g)	)			
	Reduction of Press per Mass Coal (kPa	0	0.094336833	0.164424072	0.210872837
	Reduction of Pressure(kPa)	0	6.38	11.12	14.87
	Corrected Data (kPa)	17	10.62	5.88	2 13
	Pressure (kPa)	17.26	10.88	6.14	2.39
oditiple coal size. I voulith	Time (min)	0	~	2	ო

TABLE A7: Experimental Data for the pH Condition Variation on CO<sub>2</sub> Adsorption – Untreated Conditions

 $T = 40^{\circ}C$ 

Mass of coal sample: 60g

**Reduction of Pressure** per Mass Coal (kPa/g) 0.027833333 0.188333333 0.243833333 0.255833333 0.138333333 0.054166667 0.099166667 0.156166667 0.2033333333 0.217666667 0.1195 0.173 0.2675 0.2785 0.077 0.231 Pressure(kPa) Reduction of 13.06 15.35 16.05 10.38 11.3 13.86 14.63 16.71 3.25 4.62 5.95 12.2 7.17 1.67 9.37 8.3 C Corrected Data (kPa) 13.75 12.38 11.05 15.33 0.95 7.63 1.65 0.29 9.83 6.62 3.94 3.14 2.37 5.7 4.8 8.7 ~ Pressure (kPa) 17.76 16.09 14.51 13.14 10.59 9.46 11.81 8.39 6.46 1.05 7.38 5.563.13 2.41 1.71 4.7 3.9 Sample coal size: 1000µm pH Condition: Untreated Time (min) 4 13 14 15 ÷ 16 10 თ 2 က ß ဖ ω 0 4  $\sim$ <del>.</del>

TABLE A8: Experimental Data for the pH Condition Variation on CO<sub>2</sub> Adsorption – Acidic Conditions (pH of 0.51)

T = 40°C Mass of coal sample: 80g Sample coal size: 1000µm

lition: Acidic					
in)	Pressure (kPa)	Corrected Data (kPa)	Reduction of Pressure(kPa)	Reduction of Pressure per Mass Coal (kPa/g)	
	17.95	17	0	0	
	16.14	15.19	1.81	0.022625	
	14.46	13.51	3.49	0.043625	
	12.9	11.95	5.05	0.063125	
	11.44	10.49	6.51	0.081375	
	10.1	9.15	7.85	0.098125	
	8.84	7.89	9.11	0.113875	
	7.68	6.73	10.27	0.128375	
	6.55	5.6	11.4	0.1425	
	5.48	4.53	12.47	0.155875	
	4.5	3.55	13.45	0.168125	
	3.55	2.6	14.4	0.18	
	2.67	1.72	15.28	0.191	
	1.8	0.85	16.15	0.201875	
	0.98	0.03	16.97	0.212125	

TABLE A9: Experimental Data for the pH Condition Variation on CO<sub>2</sub> Adsorption – Alkaline Conditions (pH of 12.40)

T = 40°C Mass of coal sample: 80g Sample coal size: 1000µm

г	r										-				-				
	Reduction of Pressure per Mass Coal (kPa/g)	0	0.023562677	0.044701764	0.063821193	0.081324896	0.097347516	0.112158341	0.126295947	0.138952471	0.150935775	0.16224586	0.173152013	0.183519591	0.193213949	0.202369732	0.211121583	0.219604147	0.227682779
	Reduction of Pressure(kPa)	0	1.75	3.32	4.74	6.04	7.23	8.33	9.38	10.32	11.21	12.05	12.86	13.63	14.35	15.03	15.68	16.31	16.91
	Corrected Data (kPa)	17	15.25	13.68	12.26	10.96	9.77	8.67	7.62	6.68	5.79	4.95	4.14	3.37	2.65	1.97	1.32	0.69	60.0
	Pressure (kPa)	17.81	16.06	14.49	13.07	11.77	10.58	9.48	8.43	7.49	6.6	5.76	4.95	4.18	3.46	2.78	2.13	1.5	0.9
pH Condition: Alkali	Time (min)	0		2	e	4	ŋ	9	7	œ	6	10	11	12	13	14	15	16	17

TABLE A10: Experimental Data for the pH Condition Variation on CO<sub>2</sub> Adsorption – Near Neutral Conditions (pH of 5.97)

 $T = 40^{\circ} C$ 

Mass of coal sample: 80g Sample coal size: 1000µm

pH Condition: Near - N	eutral			
Time (min)	Pressure (kPa)	Corrected Data (kPa)	Reduction of Pressure(kPa)	Reduction of Pressure per Mass Coal (kPa/g)
0	17.36	17	0	0
~	15.95	15.59	1.41	0.017625
2	14.64	14.28	2.72	0.034
3	13.38	13.02	3.98	0.04975
4	12.2	11.84	5.16	0.0645
Q	11.1	10.74	6.26	0.07825
9	10.05	9.69	7.31	0.091375
7	9.08	8.72	8.28	0.1035
8	8.15	7.79	9.21	0.115125
6	7.26	6.9	10.1	0.12625
10	6.42	6.06	10.94	0.13675
11	5.6	5.24	11.76	0.147
12	4.82	4.46	12.54	0.15675
13	4.09	3.73	13.27	0.165875
14	3.37	3.01	13.99	0.174875
15	2.69	2.33	14.67	0.183375
16	2.05	1.69	15.31	0.191375
17	1.42	1.06	15.94	0.19925
18	0.81	0.45	16.55	0.206875

## **APPENDIX B:**

## COAL ANALYSIS EXPERIMENTAL

## DATA

## &

## **ASTM STANDARDS**

## **APPENDIX B**

## LIST OF APPENDICES:

Arranged in the following order:

- 1. Results from Atomic Absorption Spectrometer (AAS)
- 2. Results from CHNS-O Analysis
- 3. ASTM Standards:
  - a. Standard Classification of Coals by Rank
  - b. Standard Test Method for Moisture in the Analysis Sample of Coal and Coke
  - c. Standard Test Method for Trace Elements in Coal and Coke Ash by Atomic Absorption

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Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

Meas. Date	e :9/14/04 4:05	PM		
Simple TD	.01 .0TD1	O a must be a bl		
sample in	:51D1	Sample Name:		
NC.		Conc.(ppm) 0.00	ABS D DDD1	REF _0_0015
Mean			0.0001	-0.0015
50			0.000	0.0000
882			0.00	0.00
Sample ID	:STD2	Sample Name:		
Ne.	Corr.Conc.	Conc (nom)	383	RFF
-		20 00		0 0016
Mean			1.3273	
50			2 2 2 2 2 2	0 0000
1. an '				· • · · · · · · · · · · · · · · · · · ·
				5 <b>• 5</b> 5
Sample 10	:STD3	Sample Name:		
Nc.	Corr.Cone.	Conc. (ppm)	ABS	REF
3		40.00	0.0913	0.0027
Mean			0.0913	0.002
SD			0.0000	0.0000
RSD(.)		'	0.00	0.00
Comple ID		(2		
Sample i	101D4 Comp Comp	Sample Name:	3.0.0	
NO.	Corr.Conc.	Conc. (ppm)	ABS	REF
1 16		80.00	0.1769	0.0067
Mean			0.1769	0.0067
SD .			0.0000	0.0000
RSD(1)			(),()()	0.00
Coefficier	)T :K3=+- K2=++	· •,		
	K1=2,201143E+0(	) २		
	KO = 1 - 860001F - 00	) 3		
Corr.Coef.	:0.9998			
	• X1 • 212121X1	<b>.</b> .		•
			•	
		·. ·		
			• .	
			·	
Sample ID	:UNK-001	Sample Name:sam	ole contai	ning coal
No.	Corr.Conc.(ppm)	Conc. (ppm)	ABS	REF
	436.34	436.34	0.9623 0	0.0500
Mean	436.34	436.34	0.9623	0.0500
SD	0.00	0.00	0.0000	0.0000
RSD (	0.00	0.00	0.00	0.00
Sample 11	• INNE-002	Sample Name+sam	sis intant	
sançı⊂ ii No	ל merer בנוכה, ייייכי	Conc (nnm)	ARS	REF
-	Sec. S	386 78	्रा दिस्टिय ्	10 0415
Ma = 11	2000 + 20 RAN 18	386 18	2 2510	3 0415
275		0.00	2.3366	0.0000
RSD	0.00	0.00	0.00	0.00

Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

ι					
STD1 orr.Conc. - - -	Sample Name Conc.(ppm) 0.00  	2: 7	ABS -0.0002 -0.0002 c.0000 c.000	REF -0.0061 -0.0061 0.0000 0.00	
STD2 orr.Conc. - - -	Sample Name Conc.(ppm) 10.00  	2: _P	ABS 0.0452 0.0462 0.0000 0.000	REF 0.0002 0.0002 0.0000 0.000	
STD3 orr.Conc. - - -	Sample Nama Conc.(ppm) 20.00  	ë: F	ABS 0.0979 0.0979 0.0000 0.000	REF 0.0079 0.0079 0.0000 0.000	
STD4 orr.Conc. - - - :K3=	Sample Name Conc.(ppm) 40.00  	2: F	ABS 0.1901 0.1901 0.0000 0.00	REF 0.0181 0.0181 0.0000 0.00	
K2= K1=4.758857E-0( K0=7.200008E-0( :0.9999	03 04		•	•	•
	:Al. STD1 orr.Conc. - - - STD2 orr.Conc. - - - STD3 orr.Conc. - - - - - - - - - - - - -	:Al. STD1 Sample Name orr.Conc. Conc. (ppm)  STD2 Sample Name orr.Conc. Conc. (ppm)  STD3 Sample Name orr.Conc. Conc. (ppm)  STD4 Sample Name orr.Conc. Conc. (ppm)  STD4 Sample Name Conc. (ppm)  STD4 Sample Name Conc. (ppm)   STD4 Sample Name Conc. (ppm)   STD4 Sample Name Conc. (ppm)    STD4 Sample Name Conc. (ppm)    STD4 Sample Name Conc. (ppm)    STD4 Sample Name Conc. (ppm)     STD4 Sample Name Conc. (ppm)     STD4 Sample Name Conc. (ppm)     	:Al. STD1 Sample Name: orr.Conc. Conc.(ppm) F 0.00 	:Al.       STD1       Sample Name:       ABS         orr.Conc.       Conc.(ppm)       ABS         -       -       0.000       -0.0002         -       -       -       0.000         -       -       -       0.000         -       -       -       0.0002         -       -       -       0.0000         -       -       -       0.0000         STD2       Sample Name:       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0452         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000         -       -       0.0000     <	:Al.         STD1       Sample Name:         orr.Conc.       Conc. (ppm)         -       0.00         -       -0.0002         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         STD2       Sample Name:         orr.Conc.       Conc. (ppm)         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         -       -         STD3       Sample Name:         orr.Conc.       Conc. (ppm)         -       -         -       -         -       -         -       -         orr.Conc.       Conc. (ppm) <t< td=""></t<>

Sample	ΙD	:UNK-001	Sample Name:sample containing coal		
NC.		Corr.Conc.(ppm)	Conc. (ppm)	ABS	REF
3		155.79	155.79	0.7421 (	0.1725
Mean		155.79	155.79	0.7421	0.1725
SD		0.00	0.00	0.0000	0.0000
$RSD\left( \cdot \right)$		0.00	0.00	0.00	0.00
Sample	ID	:UNK-002	Sample Name:sample (blanks)		
No.		Corr.Conc.(ppm)	Conc.(ppm)	ABS	REF
-		126.43	126.43	0.6024 \	0.0853
Mean		126.43	126.43	0.6024	0.0853
SC		0.00	0.00	0.0000	0000.0
RSD - Y		0.00	0.00	0.00	0.00

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Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

Meas. Date	:9/14/04 10:53	AM					
Element	:Ca						
Sample IP	:STD1	Sample Name:					
No.	Cerr.Cone.	Conc. (ppm)	ABS ·	REF			
-		0.00	0.0004	-0.0067			
Mean			0.0004	-0.0067			
30			0.0000	0.0000			
RSD 1			0.00	0.00			
Sample ID	:STD2	Sample Name:					
No.	Corr.Conc.	Conc. (ppm)	ABS	REF			
-	<u></u>	0.50	0.0262	-0.0013			
Mean			0.0262	-0.0013			
SP			0 0000	0 0000			
RSDIN			2.2.2	2.22			
A C C Y							
Sample ID	:STD3	Sample Name:					
No.	Corr Cono	Cone (nom)	ARS	275			
		1 00	0.0507	- 0.082 -			
Moan			0.0507	0.0082			
uean			0.000	0.0002			
			0.0000	0.0000			
KOU(I)			0.00	0.00			
Sample ID	· STD4	Sample Name:					
No.	Corr Coro	Conce (nom)	ABS	DEE			
1		2 00	0 0847	0 0068			
L Marin		2.00	0.0047	0.0000			
Mean. SD			0.000	0.0000			
			0.0000	0.0000			
RSU(;). Coofficier			0.00	0.00			
COETTOTEL	NS=	1.1.1 <sup>1</sup>					
	$\kappa_{2} = -$ $\kappa_{1} = $	1.0					
	N174.18UD/2E-UU2 10-2.0000000 000						
$\nabla V = 0.920$							
CULT.CORT.	• 0 • 22022	•		•			
			•				
		•					
		•	<b>.</b>	11 11-11			
Sample ID	:UNK-001	Sample Name:samp	ole contai	nind coal			
No.	Corr.Conc.(ppm)	Conc. (ppm)	LBS	REF			
1	3.14	3.14	0.1352 C	0.0057			
Mean	3.14	3.14	0.1352	0.0057			
SD	0.00	0.00	0.0000	0.0000			
RSD()	0.00	0.00	0.00	0.00			
Sample 1D	:UNK-002	Sample Name:sam	ple (blank	5			
NO.	Corr.Conc.(ppm)	Conc.(ppm)	ABS	REF			
-	6.01	6.01	0.2551 0	C.CIIZ			
Mean	6.01	6.01	0.2551	0.0112			
30	0.00	0.00	9.0000	9.0000			
RSD()	6.99	0.00	0.00	0.00			
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Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

Meas. Date	:9/14/04 11:20	AM		
Ziendin IO	. INd	0		
Sample IL.	15101 Serve Oraci	Sample Name:	200	
NC.	corr.cond.	Conc.(ppm)	ABS	REF
-		0.00	+0.0002	-0.0005
Mean			-0.0002	-0.0005
SC			0.0000	0.000
380			0.00	0.00
Sample 19	:STD2	Sample Name:		
N 5		Conc (nom)	ARC	BLL
		0.20	- 1022	
Marr		0.20	1 0020	0.00
neal.			0.0000	0.0010
5.			0.0000	0.0000
RSD			(, ())	t.t.
Sample ID	:STD3	Sample Name:		
No.	Corr.Conc.	Conc. (ppm)	ABS	REF
-		0.40	0.0117	0.0024
Mean			0.0117	0.0024
SD			0 0000	0 0000
			0.0000	0.00
KSD(7)			0.00	0.00
Sample ID	:STD4	Sample Name:		
No.	Corr.Conc.	Conc. (ppm)	ABS	REF
5		0.80	0.0272	0.0065
Monan			0.0072	2 0045
Mean .			0.02.2	0.0000
22				0.0000
RSD (C.)			0.00	0.00
Coefficier	52=	Ner 2		
	K1=3.307143E-00	02		
	KO=2.999998E-00			
Corr.Coel.	. :0.9905			
		·	•	
		· - ·	•	
		1	land. Art	111
Sample ID	:UNK-001	Sample Name:s	sampie conta	ining coal
NC.	Corr.Conc.(ppm)	Conc.(ppm)	ABS	REF
-	2.95	2.95	0.0978	C 2.604 <sup>m</sup>
Mean	2.95	- 2,95	0.0978	2.6047
57	0.20	0.00	0.0000	0.000
RSD()	0.00	0.00	0.00	0.00
Samp'e IP	:UNK-002	Sample Name:s	sample (blan	ks)
N S	(שרות בווב בייניב	 (ດກວ. (ກວະ.)	ABS	REF
	a so aga	12 56	2 314A	5 7,7253
-	a da este en el compositor de la compositor	iaich Io Rai	· / · · ·	
Negli	5 55	12+5V 0.00	· · · · ·	
SE	U.UU	0.00		5.000CC 5.000
SSD (	1. CN	U.U.U.	x = x x	5 <b>- 1</b> 75

Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

Sample ID :STD1       Sample Name:         Nc.       Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.00       -0.0010       -0.0002         Mean         0.000       -0.0010       -0.0002         ST         0.000       0.0000       0.0000         SST         0.00       0.000       0.000         Sample ID :STD2       Sample Name:        0.00       0.000         SST         0.00       0.000         Sample ID :STD2       Sample Name:        0.000       0.000         SST         0.000       0.000       0.000         SST         0.100       0.0404       0.2056         Mean         0.000       0.0000       0.0000         SST         0.20       0.1360       0.0170         Mean         0.000       0.0000       0.000         SST         0.000       0.000       0.000         SST <t< th=""><th>Meas. Date Element</th><th>e :9/14/04 11:14 :Ma</th><th>AM</th><th></th><th></th></t<>	Meas. Date Element	e :9/14/04 11:14 :Ma	AM		
Ne.       Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.00       -0.0010       -0.0002         ST         0.000       0.000         Sample ID :STD2       Sample Name:        0.2000       0.0050         ST        0.10      2404       0.0050       0.0050         ST         0.2000       0.0050       0.0050         ST         0.000       0.0050       0.0050         ST         0.2000       0.1360       0.0170         Mean         0.2000       0.1360       0.0170         St         0.000       0.000       0.000         St         0.000       0.000       0.000<	Sample ID	:STD1	Sample Name:		
1        0.00       -0.0010       -0.0002         Mean         0.000       -0.0002         ST         0.00       0.000         SST         0.00       0.000         Sample ID :STD2       Sample Name:        0.00       0.000         Mean        0.10       2.2404       0.2555         Mean        0.10       2.2404       0.056         SD         0.2000       0.000         RST         0.2000       0.000         Sample ID :STD3       Sample Name:        0.2000       0.0000         RST         0.2000       0.0000       0.0000         Sample ID :STD3       Sample Name:        0.000       0.0000       0.0000         SD         0.000       0.0000       0.0000         Sample ID :STD4       Sample Name:        0.000       0.000         SD         0.000       0.000       0.000         SD         0.000       0.000 <td>No.</td> <td>Corr.Conc.</td> <td>Conc. (ppm)</td> <td>ABS</td> <td>RFF</td>	No.	Corr.Conc.	Conc. (ppm)	ABS	RFF
Mean         -0.0010       -0.0002         ST         0.0000       0.0000         SSD         0.000       0.000         Sample ID :STD2       Sample Name:        0.00       0.000         Mean        0.10       2.2404       0.2050         Mean        0.10       2.2404       0.2050         SSD         0.2000       0.2000         SSD         0.2000       0.2000         SSD         0.2000       0.2000         SSD         0.2000       0.2000         SSD         0.20       0.1360       0.0170         Mean         0.20       0.1360       0.0170         Mean         0.000       0.000       0.000         SSD(7)         0.000       0.000       0.000         SSD(7)         0.000       0.000       0.000         SSD(7)         0.000       0.000	-	<b></b>	0.00	-0.0010	-0.0002
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Mean			-0.0010	-0.0002
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	30	~-	<b>_</b>	0.0000	0.0000
Sample 1D :STD2       Sample Name: Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.10       0.2404       0.2050         Mean        0.10       0.2404       0.2050         SD         0.2000       0.2000         SD         0.2000       0.2000         RSD         0.2000       0.2000         Sample ID :STD3       Sample Name:       No.       Corr.Conc.       Conc.(ppm)         Mean        0.20       0.1360       0.0170         Mean         0.0000       0.0000         SSD(x)         0.000       0.0000         Sample ID :STD4       Sample Name:       No.       Corr.Conc.       Conc.(ppm)       ABS       REF         No.       Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.40       0.2835       0.0354         SP         0.000       0.000         RSD(y)         0.000       0.000         Coefficient :K3=        0.000       0.00	532			0.00	0.00
No. Corr.Conc. Conc. (ppm) ABS REF 1 0.10 $2.2404$ 0.2050 Mean $2.0000$ 0.0000 SD $2.0000$ 0.0000 RSD $2.0000$ 0.0000 Sample ID :STD3 Sample Name: No. Corr.Conc. Conc. (ppm) ABS REF 1 0.20 0.1360 0.0170 Mean 0.1360 0.0170 SD 0.0000 0.0000 RSD (*) 0.000 0.0000 Sample ID :STD4 Sample Name: No. Corr.Conc. Conc. (ppm) ABS REF 1 0.40 0.2835 0.0354 Mean 0.2835 0.0354 SD 0.000 0.0000 RSD (*) 0.000 0.0000 Coefficient :K3= $4$ K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	Sample ID	:STD2	Sample Name:		
1        0.10       0.20404       0.2050         Mean         0.300       0.0000         SD         0.300       0.0000         RSD         0.20       0.1360       0.0170         Sample ID       :STD3       Sample Name:        0.20       0.1360       0.0170         Mean        0.20       0.1360       0.0170         SD         0.0000       0.0000         SD         0.0000       0.0000         SSD(7)         0.0000       0.0000         SD         0.0000       0.0000         SSD(7)         0.000       0.0000         SSD         0.000       0.0000         SSD(7)         0.000       0.0000         Sample ID       :STD4       Sample Name:        0.2835       0.0354         SE         0.000       0.000       0.000         RSD(7)         0.000       0.000       0.000 <td>Nc.</td> <td>Cerr.Cone.</td> <td>Conc. (ppm)</td> <td>ABS</td> <td>REF</td>	Nc.	Cerr.Cone.	Conc. (ppm)	ABS	REF
Mean         C.0404       0.0050         SD         C.0000       0.0000         RSD         C.0000       0.0000         Sample ID :STD3       Sample Name:       No.       Corr.Conc.       Conc.(ppm)         Mean        0.20       0.1360       0.0170         Mean         0.0000       0.0000         SSD(7)         0.0000       0.0000         SD         0.1360       0.0170         SD         0.0000       0.0000         SSD(7)         0.0000       0.0000         SD         0.0000       0.0000         SSD(7)         0.000       0.000         Sample ID :STD4       Sample Name:        0.2835       0.0354         SP         0.2835       0.0354         SP         0.000       0.000         RSD(7)         0.000       0.000         Coefficient :K3=        0.000	Î.		0.10	0.0404	0.0050
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Mean		·	0.0404	0.0050
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	SD	<b>-</b> -	<b></b> .	0.0000	0.0000
Sample ID :STD3       Sample Name:         Nc.       Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.20       0.1360       0.0170         Mean         0.0000       0.0000         SD         0.000       0.0000         RSD(4)         0.000       0.0000         Sample ID :STD4       Sample Name:       No.       Corr.Conc.       Conc.(ppm)       ABS       REF         1        0.40       0.2835       0.0354         Mean        0.000       0.000       0.000         RSD(7)         0.000       0.000         RSD(7)         0.000       0.000         RSD(7)         0.000       0.000         Coefficient :K3=        0.000       0.000         K1=7.352285E-001       K0=-1.394000E-002           Corr.Coef.       :0.9942        -	RSE			0.00	0.00
Nc. Corr.Conc. Conc.(ppm) ABS REF 1 0.20 0.1360 0.0170 Mean 0.1360 0.0170 SD 0.0000 0.0000 RSD(*) 0.00 0.000 Sample ID :STD4 Sample Name: No. Corr.Conc. Conc.(ppm) ABS REF 1 0.40 0.2835 0.0354 Mean 0.2835 0.0354 SD 0.000 0.000 RSD(*) 0.000 0.0000 RSD(*) 0.000 0.000 RSD(*) 0.000 0.000 Coefficient :K3= K2= % K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	Sample ID	:STD3	Sample Name:		
1        0.201       0.1360       0.0170         Mean        0.1360       0.0170         SD        0.0000       0.0000         RSD(x)        0.000       0.000         Sample ID :STD4       Sample Name:       0.00       0.00         Sample ID :STD4       Sample Name:       0.00       0.00         Mean        0.40       0.2835       0.0354         Mean        0.000       0.000       0.000         RSD(x)         0.000       0.000         SD        0.40       0.2835       0.0354         Mean         0.000       0.000         RSD(x)         0.000       0.000         RSD(y)         0.000       0.000         Coefficient       :K3=        0.000       0.00         K0=-1.394000E-002             Corr.Coef.       :0.9942	No.	Corr.Cond.	Conc.(ppm)	ABS	REF
Mean         0.1360       0.0170         SD         0.0000       0.0000         RSD(x)         0.00       0.000         Sample ID :STD4       Sample Name:        0.00       0.00         Sample ID :STD4       Sample Name:        0.00       0.00         Mean        0.40       0.2835       0.0354         SD        0.0000       0.0000       0.0000         RSD(x)         0.000       0.0000         SD         0.000       0.0000         Coefficient :K3=        0.000       0.000         K2=         0.000       0.000         K0=-1.394000E-002             Corr.Coef.       :0.9942	1		0.20	0.1360	0.0170
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Mean			0.1360	0.0170
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	SD			0.0000	0,0000
Sample ID :STD4     Sample Name:       No.     Corr.Conc.     Conc.(ppm)       1      0.40       Mean      0.2835       SD      0.2835       SD      0.2835       Coefficient :K3=        K2=     W       K1=7.352285E-001     W       K0=-1.394000E-002     W	$RSD\left(\mathbb{P} ight)$			0.00	0.00
No. Corr.Conc. Conc. (ppm) ABS REF 1 0.40 0.2835 0.0354 Mean 0.2835 0.0354 SD 0.000 0.0000 RSD(V) 0.000 0.000 Coefficient :K3= K2= K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	Sample ID	:STD4	Sample Name:		
1        0.40       0.2835       0.0354         Mean        0.2835       0.0354         SD        0.0000       0.0000         RSD(%)        0.000       0.000         Coefficient       :K3=       W         K1=7.352285E-001       W       W         K0=-1.394000E-002       W       W	No.	Corr.Conc.	Conc.(ppm)	ABS	REF
Mean 0.2835 0.0354 SD 0.000 0.0000 RSD(%) 0.00 0.000 Coefficient :K3= K2= 0.00 0.00 K0=-1.394000E-002 Corr.Coef. :0.9942	1		0.40	0.2835	0.0354
SD 0.0000 0.0000 RSD(%) 0.00 0.000 Coefficient :K3= K2= K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	Mean			0.2835	0.0354
RSD(v) 0.00 0.00 Coefficient :K3= K2= W K0=-1.394000E-002 Corr.Coef. :0.9942	SD			0.0000	0.0000
Coefficient :K3= K2= K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	RSD( )			0.00	0.00
K2= K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942	Coefficier	:t :K3=			
K1=7.352285E-001 K0=-1.394000E-002 Corr.Coef. :0.9942		N2=	·:*•		
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Corr.Coef. :0.9942 · · · ·		KO=-1.394000E-0	02		
•	Corr.Coef.	:0.9942			
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Sample ID	:UNK-001	Sample Name:sa	mple contai	ning coal
No.	Corr.Conc.(ppm)	Conc.(ppm)	ABS	REF
1	1.31	1.31	0.9483 C	0.8499
Mean	1.31	-1.31	0.9483	0.8499
SD	0.00	0.00	0.000	0.0000
RSD	0.00	0.00	0.00	0.00
Sample ID	:UNK-002	Sample Name:sa	ample (blank	
Sample IP No.	:UNK-002 Corr.Conc.(ppm)	Sample Name:sa Conc.(ppm)	ample (blank ABS	s) REF
Sample ID No. I	:UNK-002 Corr.Conc.(ppm) 1.98	Sample Name:sa Conc.(ppm' 0.98	ample (blank ABS 2.7053 c	REF 1.1522
Sample IP No. I Mean	:UNK-002 Corr.Conc.(ppm <sup>*</sup> 1.98 0.95	Sample Name:sa Conc.(ppm' 0.98 0.96	ample (blank ABS 0.7053 0 0.7053	REF 1.1522 1.1522
Sample IP No. I Mean SD	:UNK-002 Corr.Conc.(ppm <sup>3</sup> 1.48 0.95 0.00	Sample Name:sa Conc.(ppm' 0.98 0.98 0.00	ample (blank ABS 0.7053 0 0.7053 0.0000	as) REF 1.1522 1.1522 0.0000

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Analysi Analysi Comment	s M s N	lode :Flame/Autosa Jame :Metal conter :Suhana Muhan	ampler it in coal sample nad	analysis		
Meas. D	ate	e :9/14/04 11:0	)7 AM			
Element		:K				
Sample	ID	:STD1	Sample Name:			
No.		Corr.Conc.	ິດກຸດ (ຫຼາຫຼ)	AR3	375	
1			0.00	0 100-	·····	
Moan				0.000-	· · · · · · · · · · · · · · · · · · ·	
rean				0.000	<pre></pre>	
SU DCD INN				0.0000		
RSD())				10.00		
_ 、						
Sample	ΙD	:STD2	Sample Name:			
No.		Corr.Conc.	Conc.(ppm)	ABS	REF	
-			0.25	0.0431	0.0054	
Mean				0.0431	0.0084	
SD			Angen Witten	0.000	0.0000	
222 1				1 00	0.00	
116.6				( • C)(	U. • U.U.	
Comple	т ¬		Dama I. Mama .			
Sampre	ЦЦ	15105	sampie Name:			
NC.		Corr.Conc.	conc. (ppm)	ABS	REF	
1			0.50	0.0625	0.0109	
Mean				0.0625	0.0109	
SD				0006.9	0.0000	
RSD ( ; )				0.00	0.0Č	
Sample	TD	:STD4	Sample Name.			
No	T C.	Corr Conc	Conc (ppm)	ABS	DEE	
1		COLL.CONC.		20000	NET OTES	
1 .			Τ. Ο()	0.0908	0.0100	
Mean				0.0968	0.0.22	
SD				0.0000	0.0000	
$RSD(\cdot)$				0.00	0.00	
Coeffic	ier	nt :K3=				
		K2=				
		K1=9.137143E-0				•
		K0=1.080000E-0	002			
Corr.Co	et.	:0.9730			•	
0012100						
			•		• · · · · ·	
					1. <u>* </u> * 1	
Sample	10	:UNK-001	Sample Name:sam	ncie conta	ining coal	
No.		Corr.Conc.(ppm)	Conc.(ppm) .	, ABS	REF	
1		12.40	12.40	1.1442	C 0.3195	
Mean		12.40	12.40	1.1442	0.3195	
50		0.00	0.00	0.0000	0.0000	
830		a ee	8.00	2.00	0.00	
inch i						
Samola	T EN	·INK-002	Sample Name:sam	nole (blan	(ks)	
No.		Contraction (Contraction)	Conic (nom)	1885 1885	REF	
NC .			to at	1 32.13	- 0 1 T G C	
				0 4 7 7 4 C 10 - AZ 1 7	s s.e. 4 S.e.t. <u>5</u> T	
Mean				1.1 M D 2 D	N 4 4 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	
SE			i i i i			
RSD		0.00	0.00	· • · · ·	()(	

Analysis Mode :Flame/Autosampler Analysis Name :Metal content in coal sample analysis Comment :Suhana Muhamad

Meas. Date	· :9/14/04 11:00	AM			
Sample ID No. 1 Mean SD RSD(1)	:STD1 Corr.Conc.	Sample Name Conc.(ppm) 0.00  	2 <b>:</b>	ABS -0.0003 -0.0003 0.0000 0.00	REF 0.0000 0.0000 0.0000 0.000
Sample ID No. 1 Mean SD RSD(>)	:STD2 Corr.Conc.   	Sample Name Conc.(ppm) 1.00  	2:	ABS 0.0304 0.0304 0.0000 0.00	REF 0.0010 0.0010 0.0000 0.000 0.00
Sample ID No. 1 Mean SD RSD(>)	:STD3 Corr.Conc.	Sample Name Conc.(ppm 2.00  	2: )	ABS 0.0523 0.0523 0.0000 0.00	REF 0.0024 0.0024 0.0000 0.00
Sample ID No. 1 Mean SD RSD(%) Coefficien Corr.Coef.	:STD4 Corr.Conc.   at :K3= K2= K1=3.435714E-00 K0=-5.100001E-0 . :0.9913	Sample Nam Conc.(ppm 4.00   	e: ) Aug	ABS 0.1377 0.1377 0.0000 0.00	REF 0.0073 0.0073 0.0000 0.00
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Sample	IΓ	:UNK-001	Sample Name:sar	nple contai	ining coal
No.		Corr.Conc.(ppm)	Conc. (ppm)	ABS	REF
1		23.94	23.94	0.8175 (	0.1356
Mean		23.94	23.94	0.8175	0.1356
SD		0.00	0.00	0.0000	0.0000
RSD(1)		0.00	0.00	0.00	0.00
Sample	ID	:UNK-002	Sample Name:sam	mple (blani	KS]
No.		Corr.Conc.(ppm)	Conc. (ppm)	ABS	REF
-		6.34	5.34	0.1782 0	0.0128
Mean		5.34	5.34	0.1782	0.0128
SD		0.00	0.00	0.0000	0.0000
$RSD_{A}$ )		0.00	0.00	0.00	0.00

COAL 43 Type 56.31% H = 4.071% .00 Peak= 3836 Time = 180 Peak= 1861 160 Blk =0.008 Calib=1.0647 Blk =0.088 COAL 44 Type 58.46% H == 4.285% .00 Peak= 3948 Time = 180 Peak= 2006 160 Blk =0.008 Calib=1.0647 Blk =0.088 COAL 45 Type 57.74% H = 4.330% 100 Peak= 3766 Time = 180 Peak= 2004 150 Blk =0.008 Calib=1.0647 Blk =0.088 COAL 46 Type **56. 17% H = 3. 909%** 100 Peak= 2927 time = 180 Peak= 1868 160 Blk =0.006 Calib=1.0647 Blk =0.089

COAL 47 Type 60.24% H = 4.200% 100 Peak= 4278 Time = 180 Peak= 2051 460 Blk =0.008 Calib=1.0647 Blk =0.088 Calib=0.9934 Blk =0.038 Calib=1.0843 Blk =0.001

WT=0.583 mg Chan # 1 N = 1.900% S = 0.364% Time = 100 Peak= 380 Time = 150 Peak= 74 09:31 Oct-08-04 Calib=0.9934 Blk =0.038 Calib=1.0843 Blk =0.001 WT=0.603 mo Chan # 1 N == 1.957% S = 0.402% Time = 100 Peak= 385 Time = 150 Peak= 43 Calib=0.9934 Blk =0.038 Calib=1.0843 Blk =0.001 WT=0.581 mg Chan # 1 1.922% S = 0.379% N ₩ Time = 100 Peak= 369 Time = 150 Peak= 32 Calib=0.9934 Blk =0.038 Calib=1.0843 Blk =0.001 WTHD.604 mg that a s  $S = 0.300 \times$ N = 1.888% Time = 150 Peak= 26 Time = 100 Peak= 374 Calib=0.9934 Blk =0.038 Calib=1.0843 Blk =0.001 WT=0.602 mg Chan # 1 N = 1.889% 8 = 0.279% Time = 100 Peak= 381 Time = 150 Fears 26

09:34 Oct-08-04 09:37 Oct-08-04 221:41 ①41七一〇8一〇4 09:44 Oct-08-04

Zaaba Mohammad

11.50*0* Chemical & HSE Technologist Chemical Engineering Programme Universiti Teknologi PETRONAS 31750 Tronoh Perak Darul Ridzuan, MALAYSIA



## Standard Classification of Coals by Rank<sup>1</sup>

This standard is issued under the fixed designation D 388; the number immediately following the designation indicates the year of original adoption or, in the case of revision, the year of last revision. A number in parentheses indicates the year of last reapproval. A superscript epsilon ( $\epsilon$ ) indicates an editorial change since the last revision or reapproval.

### 1. Scope

1.1 This standard covers the classification of coals by rank, that is, according to their degree of metamorphism, or progressive alteration, in the natural series from lignite to anthracite.

1.2 This classification is applicable to coals that are composed mainly of vitrinite.

NOTE 1—Coals rich in inertinite or liptinite (exinite), or both, cannot be properly classified because, in those macerals, the properties that determine rank (calorific value, volatile matter, and agglomerating character) differ greatly from those of vitrinite in the same coal. Often such coals can be recognized by megascopic examination. In North America, these coals are mostly nonbanded varieties that contain only a small proportion of vitrain and consist mainly of attrital materials. The degree of metamorphism of nonbanded and other vitrinite-poor coals can be estimated by determining the classification properties of isolated or concentrated vitrinite fractions, or by determining the reflectance of the vitrinite (see Test Method D 2798 and Appendix X1 of this classification). However, in the use of these vitrinite-poor coals, some properties normally associated with rank, such as rheology, combustibility, hardness, and grindability (as well as the rank determining properties) may differ substantially from those of vitrinite-rich coals of the same degree of metamorphism.

The precision of the classification of impure coal may be, impaired by the effect of large amounts of mineral matter on the determination of volatile matter and calorific value, and on their calculation to the mineral-matter-free basis.

1.3 The values stated in British thermal units per pound are to be regarded as the standard. The SI equivalents of British thermal units per pound are approximate. All other values in SI units are to be regarded as standard.

### 2. Referenced Documents

- 2.1 ASTM Standards:
- D 121 Terminology of Coal and Coke<sup>2</sup>
- D 720 Test Method for Free-Swelling Index of Coal<sup>2</sup>

D 1412 Test Method for Equilibrium Moisture of Coal at 96 to 97 Percent Relative Humidity and 30°C<sup>2</sup>

- D 1757 Test Method for Sulfur in Ash from Coal and Coke<sup>2</sup>
- D 2013 Method of Preparing Coal Samples for Analysis<sup>2</sup>
- D 2234 Practice for Collection of a Gross Sample of Coal<sup>2</sup>

- D 2798 Test Method for Microscopical Determinatio the Reflectance of Vitrinite in a Polished Specime Coal<sup>2</sup>
- D 3172 Practice for Proximate Analysis of Coal and C
- D 3173 Test Method for Moisture in the Analysis Samp Coal and Coke<sup>2</sup>
- D 3174 Test Method for Ash in the Analysis Sample of and Coke from Coal<sup>2</sup>
- D 3175 Test Method for Volatile Matter in the Ana Sample of Coal and Coke<sup>2</sup>
- D 3177 Test Methods for Total Sulfur in the Ana Sample of Coal and Coke<sup>2</sup>
- D 3180 Practice for Calculating Coal and Coke Anal from As-Determined to Different Bases<sup>2</sup>
- D 3302 Test Method for Total Moisture in Coal<sup>2</sup>
- D 4596 Practice for Collection of Channel Samples of in the Mine<sup>2</sup>
- D 5192 Practice for the Collection of Coal Samples Core<sup>2</sup>
- D 5865 Test Method for Gross Calorific Value of Coal Coke<sup>2</sup>

### 3. Terminology

3.1 *Definitions*—For additional definitions of terms us this classification, refer to Terminology D 121.

3.1.1 agglomerating, adj—as applied to coal, the proj of softening when it is heated to above about 400°C nonoxidizing atmosphere, and then appearing as a coh mass after cooling to room temperature.

3.1.2 apparent rank, n—of coal, the rank designation tained on samples other than channel samples or core san with 100 % recovery, but otherwise conforming to procee of Classification D 388.

3.1.3 coal seam, n—the stratum, layer, or bed of coal lies between two other rock layers whose compositions ( significantly from that of coal.

3.2 Abbreviations—Where it is desired to abbreviate designation of the ranks of coal, the following abbrevia shall be used:

ma—meta-anthracite an—anthracite sa—semianthracite Ivb—low volatile bituminous mvb—medium volatile bituminous ftvAb—high volatile A bituminous hvBb—high volatile B bituminous hvCb—high volatile C bituminous

<sup>&</sup>lt;sup>1</sup> This classification is under the jurisdiction of ASTM Committee D-5 on Coal and Coke and is the direct responsibility of Subcommittee D05.18 on Classification of Coals.

Current edition approved Sept. 10, 1999. Published January 2000. Originally published as D 388 - 34 T. Last previous edition D 388 - 98a.

<sup>&</sup>lt;sup>2</sup> Annual Book of ASTM Standards, Vol 05.06.

JbA--subbituminous A bB--subbituminous B bC--subbituminous C JA--lignite A B--lignite B

### gnificance and Use

This classification establishes categories of coal based adational properties that depend principally on the degree letamorphism to which the coal was subjected while d. These categories indicate ranges of physical and ical characteristics that are useful in making broad lates of the behavior of coal in mining, preparation, and

### asis of Classification

Classification is according to fixed carbon and gross ific value (expressed in British thermal units per pound) lated to the mineral-matter-free basis. The higher-rank are classified according to fixed carbon on the dry basis; ower-rank coals are classified according to gross calorific on the moist basis. Agglomerating character is used to rentiate between certain adjacent groups.

### lassification by Rank

Fixed Carbon and Gross Calorific Value—Coàls shall assified by rank in accordance with Table 1. Classify coals ug gross calorific values of 14 000 Btu/lb or more on the t, mineral-matter-free basis, and coals having fixed carbon % or more on the dry, mineral-matter-free basis, accordto fixed carbon on the dry, mineral-matter-free basis. sify coals having gross calorific values less than 14 000 lb on the moist, mineral-matter-free basis according to s calorific value on the moist, mineral-matter-free basis, ided the fixed carbon on the dry, mineral-matter-free basis ss than 69 %.

2 Agglomerating Character—Classify coals having 86 % lore fixed carbon on the dry, mineral-matter-free basis, if omerating, in the low volatile group of the bituminous 2. Classify coals having gross calorific values in the range 1 10 500 to 11 500 Btu/lb on the moist, mineral-matter-free 3 according to their agglomerating character (Table 1).

<sup>3</sup> Supplemental Information—A correlation of the ranking erty, volatile matter (100—fixed carbon), with the meanimum reflectance of the vitrinite group macerals in coals d in one laboratory over a period of several years is shown ppendix X1.

### ampling

1 Samples—Classify a coal seam, or part of a coal seam, 19 locality based on the average analysis and gross calorific e (and agglomerating character where required) of not less three and preferably five or more face channel samples or samples taken in different and uniformly distributed lities, either within the same mine or closely adjacent es representing a continuous and compact area not greater approximately four square miles in regions of geological ormity. In regions in which conditions indicate that the coal ably varies rapidly in short distances, the spacing of pling points and grouping of analyses to provide average values shall not be such that coals of obviously different rank will be used in calculating average values.

7.1.1 Take channel samples by excluding mineral partings more than 1 cm ( $\frac{3}{8}$  in.) and lenses or concretions (such as sulfur balls) more than 1.25 cm ( $\frac{1}{2}$  in.) thick and 5 cm (2 in.) wide, as specified in Practice D 4596.

7.1.2 A drill core sample may be used provided it was collected as specified in Practice D 5192 and meets the following provisions: core recovery is 100 % of the seam, the major mineral partings and concretions are excluded as specified in 7.1.1, and drilling mud is removed from the core (see also 7.1.6).

7.1.3 Place all samples in metal or plastic cans with airtight lids, or heavy vapor impervious bags, properly sealed to preserve inherent moisture.

7.1.4 Analyses of samples from outcrops or from weathered or oxidized coal shall not be used for classification by rank.

7.1.5 In case the coal is likely to be classified on the *moist* basis, that is, inclusive of its natural complement of inherent moisture, take samples in a manner most likely to preserve inherent moisture for purposes of analysis. Because some of the moisture in a freshly collected sample condenses on the inside of the sample container, weigh both the container and the coal before and after air drying, and report the total loss in weight as air-drying loss.

7.1.6 If the sample is a core or if it is impossible to sample the coal without including visible surface moisture, or if there may be other reasons to question the accuracy of inherent moisture content determinable from the sample, and the coal is likely to be classified on the *moist* basis, the sampler shall include the following statement in the description: *Moisture questionable*. Samples so marked shall not be used for classification on a moist basis unless brought to a standard condition of moisture equilibrium at 30°C in a vacuum desiccator containing a saturated solution of potassium sulfate (97 % humidity) as specified in Test Method D 1412. Analyses of such samples that have been treated in this manner shall be designated as *samples equilibrated at 30°C and 97 % humidity*.

7.2 Other Types of Samples—A standard rank determination cannot be made unless samples have been obtained in accordance with 7.1. However, the relation to standard determinations may be usefully given for other types of samples taken under unspecified conditions, providing the same standards of analysis and computation are followed. Designate these comparative indications as *apparent rank*, which indicates the correct relative position for the sample analyzed but does not imply any standards of sampling. Whenever apparent rank is stated, give additional information as to the nature of the sample.

7.2.1 The apparent rank of the coal product from a mine shall be based on representative samples taken in accordance with the Organization and Planning of Sampling Operations section (Section 7) of Practice D 2234.

7.2.2 In case the coal is likely to be classed on the *moist* basis, take samples at the tipple or preparation plant and seal the sample to prevent loss of moisture.

### 8. Methods of Analysis and Testing

8.1 Laboratory Sampling and Analysis-Prepare coal in

accordance with Method D 2013 and analyze it in accordance with Test Methods D 3173, D 3174, D 3175, D 3177, D 3302, and Practice D 3172. Determine its gross calorific value in accordance with Test Method D 5865. Determine the sulfur trioxide (SO<sub>3</sub>) retained in the ash in accordance with Test Method D 1757 and express the result on a dry basis. Inherent moisture is reported as as-received moisture if the sample was collected according to 7.1.1 or as equilibrium moisture if 7.1.6 (Test Method D 1412) applies.

8.2 Adjust the ash value determined in accordance with Test Method D 3174 to be free of sulfate as follows:

$$A = A_{c} \left( 1 - \frac{SO_{3}}{100} \right) \left( 1 - \frac{M}{100} \right)$$
(1)

where:

- A = adjusted ash value on the inherent moist basis,
- $A_d$  = ash yield, dry basis, determined in accordance with Test Method D 3174,
- $SO_3 = in$  the ash determined in accordance with Test Method D 1757, and

M = inherent moisture.

Add to the value of fixed carbon that is determined in accordance with Practice D 3172 the value of the  $SO_3$  determined in the ash to obtain the value FC to be used in Eq 2.

8.3 Agglomerating Character—The test carried out by the examination of the residue in the platinum crucible incident to the volatile matter determination shall be used.<sup>3</sup> Coals which, in the volatile matter determination, produce either an agglomerate button that will support a 500-g weight without pulverizing, or a button showing swelling or cell structure, shall be considered agglomerating from the standpoint of classification. : In addition, a result of 1.0 or more on the Free Swelling Index

## 御 D 388

test (Test Method D 720) may also be used to indicate the is agglomerating; a result of 0.5 or 0 indicates the c nonagglomerating.

## 9. Calculation to Mineral-Matter-Free Basis

9.1 Calculation of Fixed Carbon and Calorific Value classification of coal according to rank, calculate fixed c and gross calorific value to the mineral-matter-free (Mrr basis in accordance with the Parr formulas,<sup>4</sup> Eq 2-4. ground infromation concerning the development of the formulas as well as other ranking considerations and exa of the calculations (Table 1) are provided in Appendix 2

9.2 Calculate to Mm-free basis as follows:

9.2.1 Parr Formulas:

Dry, Mm-free FC = 100(FC - 0.15S)/(100 - (M + 1.08A + 0.5))

Moist, Mm-free Btu = 100(Btu - 50S)/(100 - (1.08A + 0.55))

where:

Btu = gross calorific value, Btu/lb, FC = fixed carbon, %,

VM = volatile matter, %,

M = moisture, %,

$$A = ash. \%$$
 and

S =sulfur, %.

In Eq 2 and Eq 4, the quantities are all on the inh moisture basis. Fixed carbon (FC) and ash (A) are adjusted to the  $SO_3$ -free basis in accordance with 8.2.

### 10. Keywords

10.1 anthracite; bituminous; coal; lignite; rank

<sup>4</sup> Parr, S. W., "The Classification of Coal," Bulletin No. 180, Engin Experiment Station, University of Illinois, 1928.

### APPENDIXES

### (Nonmandatory Information)

# X1. CORRELATION OF VOLATILE MATTER WITH MEAN-MAXIMUM REFLECTANCE OF VITRINITE

X1.1 The reflectance of vitrinite in a sample of coal, as determined by Test Method D 2798, provides a useful guide to the rank of the coal. The correlation of the mean-maximum reflectance of all varieties of vitrinite with volatile matter, expressed on a dry and mineral-matter-free basis, is given in Fig. X1.1. Data are plotted for 807 coal samples that contained less than 8% ash from many different coal fields in North America. All data were determined by a single laboratory, with several different analysts over a period of several years. The plot shows a range of reflectances for three important rank groups:

Mean-Max, %	Distribution Midpoints	Rank
<1.15 1.02–1.55	<1.1 1.10–1.45	hvb
1.35~2.0(?)	1.45-2.0(?)	lvb

Note X1.1—Coals with the same vitrinite reflectance and sit maceral compositions may have different rheological and fluoresc properties and even burn and carbonize differently. These differences be due to such diverse factors as their geologic age, environment, ar mode of accumulation (time, temperature, and pressure), and a differences in the plants that contributed to their formation. Thus, the of vitrinite reflectance for selecting coals for use may need additi

<sup>&</sup>lt;sup>3</sup> Gilmore, R. E., Connell, G. P., and Nicholls, J. H. H., "Agglomerating and Agglutinating Tests for Classifying Weakly Caking Coals," *Transactions*, American Institute of Mining and Metallurgical Engineers, Coal Division, Vol 108, 1934, p. 255.



ications to predict their utilization potential. This is particularly tant in selecting coals for coke production since vitrinites with the

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same reflectance but different fluorescence properties are known to produce different carbon forms that have different physical (strength) and chemical (reactivity) properties.

X1.2 The midpoints given above are the midpoints of the distribution for the lower and upper boundary points on the reflectance scale for the indicated rank. Of the 807 coals, those that contain greater than 25 volume % inertinites tend to plot on the lower side of the distribution range than do the others that contain more vitrinites and liptinites.

### X2. BACKGROUND INFORMATION ON THE PARR EQUATIONS AND OTHER RANKING CONSIDERATIONS

2.1 Introduction—Coals are ranked according to Classiion D 388 on a mineral matter-free basis, dry or moist, inding on the parameter that applies. The rank meters—either volatile matter (or fixed carbon) or gross rific values—are commonly reported by laboratories on the ceived, dry-and-ash-free basis. These reported values be converted to the mineral-matter-free basis for ranking oses. Thus converted, the properties of the maceral (cariceous) material are used as ranking criteria, and the effects ariable mineral matter contents, which are unrelated to , are eliminated. In essence, only the "pure coal" fraction given sample is being ranked. The Part formula is used to nate the original mineral matter in the coal by using the ash 1 and total sulfur content determined on that coal as ws:

$$Mm = 1.08A + 0.55S$$

re:

= ash yield (ASTM Test Method D 3174) and

= total sulfur content (ASTM Test Methods D 3177).

This formula assumes that clay minerals, with an average water of hydration content of 8 %, and pyrite, which contains essentially all the sulfur, are the only mineral groups present. Furthermore, the following reactions are assumed to occur during ashing: (1) the hydroxyl groups from the clay minerals are lost to the atmosphere; (2) the sulfur converts to sulfur dioxide, which also is lost; and (3) pyrite decomposes to iron oxide and iron is retained in the ash. The Parr formula attempts to correct the measured ash and sulfur for these reactions by adjusting their mass back to that of the original minerals in the coal. By using this formula, the varying amounts of mineral matter can be factored out of the ranking of coals. For example, Samples A and B in Table X2.1 are both ranked as Lignite A because they have similar gross calorific values when calculated to a moist, mineral-matter-free basis (gross calorific value<sub>m,mmf</sub>), in contrast to their gross calorific values which are quite different on an as-received basis. In this example, differing mineral contents are thus factored out for the purposes of ranking.

Section X2.3 provides useful equations that enable the

## Standard Test Method for Moisture in the Analysis Sample of Coal and Coke<sup>1</sup>

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This standard is issued under the fixed designation D 3173; the number immediately following the designation indicates the year of original adoption or, in the case of revision, the year of last revision. A number in parentheses indicates the year of last reapproval. A superscript epsilon ( $\epsilon$ ) indicates an editorial change since the last revision or reapproval.

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his test method covers the determination of moisture in ysis sample of coal or coke. It is used for calculating alytical results to a dry basis. When used in conjunc-1 the air drying loss as determined in accordance with D 2013 or Practice D 346, each analytical result can be ed to an as-received basis:

his standard does not purport to address all of the oncerns, if any, associated with its use. It is the bility of the user of this standard to establish approafety and health practices and determine the applicaregulatory limitations prior to use.

### renced Documents

#### STM Standards:

5 Practice for Collection and Preparation of Coke pples for Laboratory Analysis<sup>2</sup>

 13 Method of Preparing Coal Samples for Analysis<sup>2</sup>
 30 Practice for Calculating Coal and Coke Analyses n As-Determined to Different Bases<sup>2</sup>

)2 Test Method for Total Moisture in Coal<sup>2</sup>

### mary of Test Method

Aoisture is determined by establishing the loss in of the sample when heated under rigidly controlled ms of temperature, time and atmosphere, sample and equipment specifications.

#### ificance and Use

Moisture as determined by this test method is used for ing other analytical results to a moisture free basis rocedures in Practice D 3180. Moisture percent deterby this test method may be used in conjunction with the moisture loss determined in Method D 2013 and Test D 3302 to determine total moisture in coal. Total is used for calculating other analytical results to "as d" basis using Practice D 3180. Moisture, ash, volatile and fixed carbon percents constitute the proximate s of coal and coke.

### 5. Analysis Sample

5.1 The analysis sample is that sample which has been pulverized to pass 250- $\mu$ m (No. 60) sieve as prepared in accordance with Practice D 346 or Method D 2013.

### 6. Apparatus

6.1 Drying Oven, for coal samples:

6.1.1 For determining the moisture of coal, the oven shall be so constructed as to have a uniform temperature in all parts, have a minimum of air space, and be capable of temperature regulation between the limits of 104 and 110°C. It may be of the form shown in Fig. 1. Provision shall be made for renewing the preheated air in the oven at the rate of two to four times a minute, with the air dried as defined in 7.1.

6.1.2 In the oven shown in Fig. 1, the door should contain a hole of approximately  $\frac{1}{8}$  in. (3.2 mm) in diameter near the bottom to permit a free flow of air through the oven space.

6.2 Drying Oven, for coke samples. For determining the moisture of coke, an ordinary drying oven with openings for natural air circulation and capable of temperature regulation between limits of 104 and 110°C may be used.

6.3 Capsules, with covers. A convenient form, which allows the ash determination to be made on the same sample, is a porcelain capsule,  $\frac{7}{8}$  in. (22 mm) in depth and  $\frac{13}{4}$  in. (44 mm) in diameter, or a fused silica capsule of similar shape. These capsules shall be used with a well-fitting flat aluminum cover, illustrated in Fig. 2. Platinum crucibles or glass capsules with ground-glass caps may also be used. They should be as shallow as possible, consistent with convenient handling.

### 7. Reagents

7.1 Dry Air—Air used to purge the drying oven should be dried to a moisture content of 1.9 mg/L or less. (Dew point  $-10^{\circ}$ C or less.) Any desiccant or drying method capable of achieving this degree of dryness is suitable.

7.2 Desiccants—Materials suitable for use in the desiccator may be chosen from the following:

7.2.1 Anhydrous Calcium Sulfate (0.004 mg/L).

- 7.2.2 Silica Gel.
- 7.2.3 Magnesium Perchlorate (0.0005 mg/L).
- 7.2.4 Sulfuric Acid, Concentrated (0.003 mg/L).

7.2.5 The desiccant must be kept fresh enough to assure that the air in the desiccator is dry as defined in 7.1. Values in parentheses () are literature values for the residual amount of moisture in air at equilibrium with these desiccants. (Warning:

test method is under the jurisdiction of ASTM Committee D05 on Coal <sup>1</sup> and is the direct responsibility of Subcommittee D05.21 on Methods of

It edition approved Nov. 10, 2000. Published January 2001. Originally as D 3173 - 73. Last previous edition D 3173 - 87 (1996). al Book of ASTM Standards, Vol 05.05.



NOTE 1—Details in U.S. Bureau of Mines Bulletin No. 492, 1951, p 6. FIG. 1 Moisture Oven



FIG. 2 Capsule for Use in Determining Moisture

Sulfuric acid is corrosive and can cause severe damage to eyes, skin, and clothing. Magnesium perchlorate is a strong oxidant and can react violently with organic materials.)

## 8. Procedure for Sample Passing a 250-µm (No. 60) Sieve

8.1 Heat the empty capsules under the conditions at which the sample is to be dried, place the stopper or cover on the capsule, cool over a desiccant for 15 to 30 min, and weigh. Dip out with a spoon or spatula from the sample bottle approximately 1 g of the sample. Put this quickly into the capsule, close, and weigh at once to the nearest  $\pm 0.1$  mg.

8.2 An alternative procedure for weighing the sample (more subject to error) is as follows: After transferring an amount of the sample slightly in excess of 1 g, bring to exactly 1 g in weight ( $\pm 0.5$  mg) by quickly removing the excess weight of the sample with a spatula. The utmost dispatch must be used to minimize the exposure of the sample until the weight is determined.

8.3 After removing the covers, quickly place the capsules in a preheated oven (at 104 to  $110^{\circ}$ C) through which passes a

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current of dry air. (The current of dry air is not necessar coke.) Close the oven at once and heat for 1 h. Open the cover the capsules quickly, cool in a desiccator over desic and weigh as soon as the capsules have reached room tern ture.

8.4 Use the percentage of moisture in the sample pass 250- $\mu$ m (No. 60) sieve to calculate the results of the analyses to a dry basis.

### 9. Calculations

9.1 Calculate the percent moisture in the analysis samp follows:

Moisture in analysis sample,  $\% = [(A - B)/A] \times 100$ 

where:

A = grams of sample used and

B = grams of sample after heating.

### 10. Precision and Bias

10.1 Precision—The precision of this method for the demination of residual moisture in the analysis sample of c and coke is shown in Table 1. The precision characterized repeatability  $(S_r, r)$  and reproducibility  $(S_R, R)$  is described Table A1.1 in Annex A1.

10.1.1 Repeatability Limit (r)—The value below which absolute difference between two test results of separate a consecutive test determinations, carried out on the sa sample in the same laboratory by the same operator using same apparatus on samples taken at random from a sin quantity of homogenous material, may be expected to oc with a probability of approximately 95 %.

10.1.2 Reproducibility Limit (R)—The value below wh the absolute difference between two test results, carried our different laboratories using samples taken at random from single quantity of material that is as nearly homogenous possible, may be expected to occur with a probability approximately 95 %.

10.2 *Bias*—Certified Reference Materials are not availa for the determination of bias by this test method.

10.3 An interlaboratory study, designed consistent w Practice E 691, was conducted in 1995. Twelve labs parti pated. The details of the study and supporting data are given ASTM Research Report D05-1020 filed at ASTM Headqu ters (see Note 1).

Note 1—One coke sample was analyzed in the interlaboratory stu-The average moisture of the coke sample was 0.36 %. The repeatabilimit for this sample is 0.10. The reproducibility limit for this sample 0.15. The repeatability standard deviation for this sample is 0.036. The reproducibility standard deviation for this sample is 0.053.

TABLE 1 Precision

Range	Repeatability Limit (r)	Reproducibility Limit
1.0 - 21.9 %	0.09 + 0.01 <del>7</del> ^	$0.23 + 0.02\bar{x}^{A}$

Where  $\overline{\mathbf{x}}$  is the average of two single test results.

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### ANNEX

### (Mandatory Information)

### **A1. PRECISION STATISTICS**

1.1 The precision of this test method, characterized by atability  $(S_r, r)$  and reproducibility  $(S_R, R)$  has been mined for the following materials as listed in Table A1.1.

deviation of test results obtained under repeatability conditions.

A1.3 Reproducibility Standard Deviation  $(S_R)$ —The standard deviation of test results obtained under reproducibility conditions.

## 1.2 Repeatability Standard Deviation $(S_r)$ —The standard

Parameters Used for Calculation of Precision Statement						
Material	Average	<i>S</i> ,	S <sub>R</sub>	· · /	R	
lvb	0.9683	0.0479	0.0853	0.1341	0.2388	
hvAb	1.5689	0.0569	0.1387 🥆	0.1593	0.3884	
hvAb	3.1178	0.0559	0.0977	0.1564	0.2736	
hvCb	5.4428	0.0601	0.1408	0.1682	0.3943	
hvCb	9.6392	0.0636	0.1004	0.1781	0.2811	
subB	11.9	0.1252	0.2193	0.3505	0.614	
subA	12.4556	0.0798	0.1188	0.2234	0.3326	
subA	15.1753	0.0577	0.1737	0.1615	0.4865	
subC	21.9461	0.1949	0.2843	0.5456	0.796	

TABLE A1.1 Repeatability ( $S_{r}$ , t) and Reproducibility ( $S_{rr}$ , R) Parameters Used for Calculation of Precision Statement

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## Standard Test Method for Trace Elements in Coal and Coke Ash by Atomic Absorption<sup>1</sup>

This standard is issued under the fixed designation D 3683; the number immediately following the designation indicates the year of original adoption or, in the case of revision, the year of last revision. A number in parentheses indicates the year of last reapproval. A superscript epsilon ( $\epsilon$ ) indicates an editorial change since the last revision or reapproval.

### 1. Scope

1.1 This test method<sup>2</sup> describes a procedure for the determination of beryllium, chromium, copper, manganese, nickel, lead, vanadium, and zinc in coal ash or coke ash.

NOTE 1—Although not included, this test method can be applicable to the determination of other trace elements, for example, cadmium.

1.2 This standard does not purport to address all of the safety concerns, if any, associated with its use. It is the responsibility of the user of this standard to establish appropriate safety and health practices and determine the applicability of regulatory limitations prior to use.

1.3 The values stated in SI units (IEEE/ASTM SI 10) should be regarded as the standard.

### 2. Referenced Documents

### 2.1 ASTM Standards:

- D 346 Practice for Collection and Preparation of Coke Samples For Laboratory Analysis<sup>3</sup>
- D 1193 Specification for Reagent Water<sup>4</sup>
- D 2013 Method of Preparing Coal Samples for Analysis<sup>3</sup>
- D 3173 Test Method for Moisture in the Analysis Sample of Coal and Coke<sup>3</sup>
- D 3180 Practice for Calculating Coal and Coke Analyses from As-Determined to Different Bases<sup>3</sup>
- D 5142 Test Methods for Proximate Analysis of the Analysis Sample of Coal and Coke by Instrumental Procedures<sup>3</sup> IEEE/ASTM SI 10 Standard for Use of the International System of Units (SI): The Modern Metric System<sup>5</sup>

Note 2—Method D 2013 specifies coal ground to pass through a 60-mesh (250- $\mu$ m) screen. For the trace element determinations in this test method -100-mesh (150- $\mu$ m) coal is recommended.

### 3. Summary of Test Method

3.1 Coal or coke is ashed, the ash is dissolved by mine acids, and the individual elements determined by atom absorption spectrometry.

#### 4. Significance and Use

4.1 Many trace elements occur in coal, primarily as a part the mineral matter but may also be associated with the orginatrix. Concern over release of certain trace elements to environment as a result of coal utilization has made determination of these elements an increasingly import aspect of coal analysis.

4.2 When coal ash is prepared in accordance with this method, the eight elements listed in 1.1 are quantitative retained in the ash and are representative of concentrations the coal. Concentrations of these and other elements in populant ash, industrial process ash, fly ash, and so forth, may may not be representative of total quantities in the coal.

### 5. Apparatus

5.1 Atomic Absorption Spectrometer—Any dual-channel strument using a deuterium  $(D_2)$  arc background correction other comparable simultaneous background correction system.

5.2 Muffle Furnace, with temperature control.

5.3 *Bottles*, polyethylene or polytetrafluoroethyle 125-mL capacity, with screw-cap lids, capable of withstand temperatures up to 130°C.

5.4 Volumetric Flasks, 100-mL capacity.

5.5 Steam Bath.

- 5.6 Analytical Balance, capable of weighing to 0.1 mg
- 5.7 Crucibles, 50-mL quartz or high silica.

#### 6. Reagents

6.1 Purity of Reagents—Reagent grade chemicals shal used in all tests. Unless otherwise indicated, it is intended all reagents shall conform to the specifications of the Con tee on Analytical Reagents of the American Chemical So where such specifications are available.<sup>6</sup> Other grades ma

<sup>&</sup>lt;sup>1</sup> This test method is under the jurisdiction of ASTM Committee D05 on Coal and Coke, and is the direct responsibility of Subcommittee D05.29 on Major Elements in Ash and Trace Elements in Coal.

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<sup>&</sup>lt;sup>2</sup> For information concerning experimental work on which this test method is based, see Bernas, B., "A New Method for the Decomposition and Comprehensive Analysis of Silicates by Atomic Absorption Spectrometry," *Analytical Chemistry*, ANCHA, Vol 40, 1968, pp 1682–86.

<sup>&</sup>lt;sup>3</sup> Annual Book of ASTM Standards, Vol 05.06.

<sup>\*</sup> Annual Book of ASTM Standards, Vol 11.01.

<sup>&</sup>lt;sup>8</sup> Annual Book of ASTM Standards, Vol 14.04.

<sup>&</sup>lt;sup>6</sup> Reagent Chemicals, American Chemical Society Specifications, American Society, Washington, DC. For suggestions on the testing of reagentisted by the American Chemical Society, see Analar Standards for Labor Chemicals, BDH Ltd., Poole, Dorset, U.K., and the United States Pharma and National Formulary, U.S. Pharmacopeial Convention, Inc. (USPC), Rom MD.

using the same apparatus, should not exceed the repeatability limits for more than 5% of such paired values (95% confidence level). When such a difference is found to exceed the repeatability limit, there is reason to question one, or both, of the test results. The repeatability limits determined by this method are listed in Table 2.

### TABLE 2 Repeatability

Element .	Air-Dried Coal	, ppm (µg/g)	
Elettietil	Repeatability <sup>A</sup>	Range <sup>B</sup>	
Be	0.2	0.5 to 5	
Cr	3	5 to 50	
Cu	2	5 to 50	
Mn	3	10 to 300	
Ni	1	<5	
Ni	3	5 to 30	
Pb	2	10 to 100	
v	5	10 to 100	
Zn	3	<50	
Zn	8	50 to 100	

<sup>A</sup>Values should be considered as a general guide for the kinds of repeatability and reproducibility obtainable and not necessarily as values from which no deviation is permissible.

<sup>B</sup>Approximate range of concentrations within which sample results used for statistical evaluation fall.

11.1.2 Reproducibility—The difference in absolute value of replicate determinations carried out in different laboratories, on representative samples prepared from the same bulk sample after the last stage of reduction, should not exceed the reproducibility limit for more than 5 % of such paired values (95 % confidence level). When such a difference is found to exceed the reproducibility limit, there is reason to question one,

or both, of the test results. The reproducibility limits mined by this test method are listed in Table 3.

11.2 Bias—Standard Reference Material 1632, a N<sub>i</sub> Institute of Standards and Technology (NIST) certifie sample, was analyzed with the results which are tabula Table 4.

Note 11—The NIST value for beryllium is not certified and is g informational purposes only. Values for Test Method D 3683 repre mean of the means from four separate laboratories, each of whic four replicate analyses on four separate samples of the coal stanc

### 12. Keywords

12.1 coal; coal ash; coke ash; trace elements; atom sorption spectroscopy

TABLE 3 Reproducibility							
Element	Air-Dried Coal, ppm (µg/g)						
	Reproducibility <sup>4</sup>	Range <sup>B</sup>					
Be	0.5	0.5 to 5					
Cr	5	5 to 50					
Cu	3	5 to 50					
Mn	6	<50					
Mn	20	50 to 500					
Ni	3	<5					
N	9	9 to 30					
Pb	9	10 to 100					
V	9	10 to 50					
V	20	50 to 100					
Zn	4	<50					
Zn	17	50 to 100					

<sup>A</sup>Values should be considered as a general guide for the kinds of rep and reproducibility obtainable and not necessarily as values from v deviation is permissible.

<sup>6</sup>Approximate range of concentrations within which sample results statistical evaluation fall.

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TABL	E 4	Bias
------	-----	------

				Air-Dried Coal,	ppm (µg/g)			
32	Be	Cr	Cu	Mn	Ni	Pb 30 + 9	V	$\frac{2n}{37 \pm 4}$
	$1.5^{A}$ 1.5 + 0.1	20.2 ± 0.5 20.0 ± 1	18.2 ± 2 17.4 ± 0.6	40 ± 3 43.8 ± 3	$15 \pm 1$ 14.6 ± 3	27.8 ± 4	45.7 ± 4	37 ± 4

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