

# **SLAGGING IN ENTRAINED-FLOW GASIFIERS**

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## Statement of Contribution of Collaborators

Chapters 3-8 of this thesis take the form of research papers which include work from collaborators, as detailed below. I am the sole author of all other chapters. My supervisors, Dr. Arturo Macchi of the Department of Chemical and Biological Engineering, University of Ottawa, and Dr. Edward John Anthony of CanmetENERGY, Natural Resources Canada, supervised my work during the Ph.D. program and provided editorial corrections.

For Chapter 3, I authored the paper. The modeling concept is mine. I developed the artificial neural network model, compared it to other models and applied it to case studies.

For Chapter 4, I authored 80% of the paper. The toolbox concept is mine. I developed 80% of the toolbox. Mr. Bronsch and Dr. Masset authored 20% of the paper and developed 20% of the toolbox.

For Chapter 5, I authored the paper. I assembled the CanmetENERGY slag viscosity measurement system with help from Dr. Hughes, Mr. McCalden and Dr. Lu. Dr. Ilyushechkin and I performed the viscosity measurements and slag quenching experiments. Dr. Ilyushechkin completed the electron microscopy analysis. I completed the FactSage modeling. Dr. Ilyushechkin and I analysed all the results.

For Chapter 6, Dr. Ilyushechkin authored the paper. Dr. Ilyushechkin performed the quenching experiments and completed the electron microscopy analysis. Dr. Ilyushechkin and I performed the viscosity measurements. Dr. Ilyushechkin and I analysed all the results.

For Chapter 7, I authored the paper. I performed or supervised the viscosity measurements. I completed the cup tests and FactSage modeling. I developed the characterization program and analysed all the results.

For Chapter 8, with the exception of the gasifier system description which was written by Dr. Hughes, I authored the paper. Dr. Hughes, Mr. McCalden and Dr. Lu are responsible for the design and operation of the gasifier system. I assisted in the gasifier maintenance and sample collection. Dr. Hughes and I developed the pilot testing program. Dr. Lu and I performed the surface area and density measurements. I performed the surface roughness measurements. I developed the solid sample analysis program and analysed all the results.

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## Abstract

Gasification is a flexible technology which is applied in industry for electricity generation, hydrogen production, steam raising and liquid fuels production. Furthermore, it can utilize one or more feedstocks such as coal, biomass, municipal waste and petroleum coke. This versatility, in addition to being adaptable to various emissions control technologies (including carbon capture) renders it an attractive option for years to come. One of the most common gasifier types is the entrained-flow slagging gasifier. The behaviour of inorganic fuel components in these gasifiers is still ill-understood even though it can be the determining factor in their design and operation. A literature review of inorganic matter transformation sub-models for entrained-flow slagging gasifiers is provided. Slag viscosity was identified as a critical property in the sub-models.

Slag viscosity models are only applicable to a limited range of slag compositions and conditions, and their performance is not easily assessed. An artificial neural network model was developed to predict slag viscosity over a broad range of temperatures and slag compositions. This model outperforms other slag viscosity models, resulting in an average absolute logarithmic error of 0.703 when applied to validation data. Furthermore, a toolbox was developed to assist slag viscosity model users in the selection of the best model for given slag compositions and conditions, and to help users determine how well the best model will perform. The toolbox includes a slag viscosity prediction calculator with 24 slag viscosity models, and a database of 4124 slag viscosity measurements.

Petroleum coke may be used as a fuel for entrained-flow slagging gasification. The slag viscosities of coal, petroleum coke and coal/petroleum coke blends were measured in the temperature range of 1175-1650°C. Two different viscosity measurement apparatuses were used in separate laboratories. Some viscosity measurements were repeated to test reproducibility of the results. Some effects of blending on viscosity can be explained by network former and modifier theory. Other effects are attributed to solids formation which was investigated via slag quenching experiments and phase equilibrium

predictions. Below 1300°C, vanadium, a major component of petroleum coke ashes, promotes the formation of spinel which increases slag viscosity. Unfortunately, slags containing vanadium species react readily with the crucible and spindle materials used for viscosity measurements. Interaction of vanadium-rich slags with various materials was investigated. The bulk and phase compositions of two petroleum coke slags on Al<sub>2</sub>O<sub>3</sub>, Mo, Pt and Ni supports were analysed, and kinetics of slag compositional changes at 1400 °C were determined. Mechanisms of the slag interactions with supports are described. Viscosity of slags with Mo, Ni and Al<sub>2</sub>O<sub>3</sub> supports are compared in the temperature range 1200-1500 °C.

The results from the first two parts of a three-part research program which involves fuel characterization, testing in a 1 MW<sub>th</sub> gasifier, and computational fluid dynamics (CFD) modeling for entrained-flow slagging gasification are presented. The end goal is to develop a CFD model which includes inorganic matter transformations. Initially, four coals were selected for this program and one limestone was chosen as a fluxing agent. Fuel properties were determined with prioritization based on their application; screening of potential fuels, ensuring proper gasifier operation, gasifier design and/or CFD modeling. Of the four coals tested, one was deemed unsuitable based on initial screening tests. Two of the three remaining coals require fluxing for proper gasifier operation. Using CanmetENERGY's 1 MW<sub>th</sub> gasifier, five gasification tests were completed with the characterized coals. Solid samples from the refractory liners, in-situ gas sampling probe sheaths and impingers, the slag tap, the slag pot, quench discharge water and scrubber water were collected and characterized. Char and fly ash samples indicate that most of the inorganic matter melted and formed spheres. Devolatilization and interaction with the gasifier refractory affected the composition of slag collected from the gasifier. Signs of refractory spalling/erosion were detected. The slag layers formed on the alumina liners are smooth with some rivulets and spotting. The slag layers formed on the alumina-chromia liners are rough and bubbly. Slag penetration fractions were determined for all liners.

## Sommaire

La gazéification est une technologie flexible qui est appliquée en industrie pour la génération d'électricité, la production d'hydrogène, la formation de vapeur et la production de carburants liquides. En outre, elle peut consommer un ou plusieurs types de carburants comme le charbon, la biomasse, les déchets municipaux et le coke de pétrole. Cette polyvalence, en plus d'être adaptable à différentes technologies pour contrôler les émissions (y compris le dioxyde de carbone) rend la gazéification une option attrayante pour les années à venir. L'un des types les plus courants de gazéificateur est le gazéificateur à flux entraîné. Le comportement de matières inorganiques dans ces gazéificateurs est encore mal compris, malgré le fait que ça peut être le facteur déterminant dans la conception et le fonctionnement des gazéificateurs. Une revue de la littérature sur les sous-modèles pour les transformations de la matière inorganique dans les gazéificateurs à flux entraîné est fournie. La viscosité du mâchefer a été identifiée comme une propriété essentielle pour les sous-modèles.

Les modèles de viscosité pour le mâchefer ne sont applicables qu'à des compositions et conditions limitées. De plus, leur performance n'est pas facile à évaluer. Un modèle à réseau neuronal artificiel, applicable à une large plage de températures et compositions, a été développé pour prédire la viscosité du mâchefer. Ce modèle surpasse autres modèles de viscosité, résultant en une erreur logarithmique absolue moyenne de 0,703 lorsqu'il est appliqué à des données de validation. De plus, une boîte à outils a été conçue pour assister les utilisateurs de modèles de viscosité du mâchefer. La boîte d'outils permet la sélection du meilleur modèle pour des compositions et conditions d'intérêts par prédiction de la performance de divers modèles. La boîte à outils comprend un calculateur de prédictions avec 24 modèles, et une base de données avec 4124 mesures.

Le coke de pétrole peut être utilisé comme carburant pour la gazéification à flux entraîné. La viscosité de mâchefers provenant de charbon, coke de pétrole, et mélanges de charbon et coke de pétrole a été mesurée dans la gamme de température de 1175-1650 °C. Deux

différents appareils de mesure de viscosité ont été utilisés dans des laboratoires indépendants. Certaines mesures de viscosité ont été répétées pour vérifier la reproductibilité des résultats. Certains effets sur la viscosité du mâchefer résultant de l'ajout du coke de pétrole au charbon peuvent être expliqués par la théorie de formeurs et modificateurs de réseaux dans les fondus de silices. D'autres effets sont attribués à la formation de matières solides qui a été étudiée par l'analyse de mâchefers refroidis, ainsi que par prédictions d'équilibre de phases. En dessous de 1300 °C, le vanadium, un composant majeur des cendres du coke de pétrole, favorise la formation de spinelle qui augmente la viscosité du mâchefer. Malheureusement, les mâchefers contenant du vanadium réagissent aisément avec les matériaux de creuset utilisés pour la mesure de viscosité. Les interactions de divers matériaux avec mâchefers riches en vanadium ont été étudiées. Les compositions moyennes et de phases de deux mâchefers provenant de cokes de pétrole placés sur du Al<sub>2</sub>O<sub>3</sub>, Mo, Pt ou Ni ont été analysées. La cinétique des changements de compositions pour les mâchefers à 1400 °C a été déterminée. Les mécanismes d'interactions avec les matériaux sont décrits. Une comparaison de la viscosité du mâchefer mesurée à l'aide de creusets en Mo, Ni et Al<sub>2</sub>O<sub>3</sub> est effectuée dans l'intervalle de température de 1200 à 1500 °C.

Les résultats des deux premiers volets d'un programme de recherche qui comprend la caractérisation de carburants, des essais avec un gazéificateur de 1 MW<sub>th</sub>, et la mécanique des fluides numérique (MFN) pour la gazéification à flux entraîné sont présentés. L'objectif final est de développer un modèle MFN de gazéification qui prend compte des transformations de la matière inorganique. Initialement, quatre charbons ont été choisis pour ce programme et un calcaire a été choisi comme agent fluxant. Les propriétés des carburants ont été déterminées avec une hiérarchisation en fonction de leur application; le dépistage des carburants potentiels, assurer un bon fonctionnement du gazéificateur, la conception du gazéificateur, et/ou la modélisation MFN. Parmi les quatre charbons choisis, l'un a été éliminé en tant que carburant potentiel. Deux des trois charbons restants nécessitent un agent fluxant pour le bon écoulement du mâchefer. En utilisant le gazéificateur de 1 MW<sub>th</sub> à CanmetÉNERGIE, cinq essais de gazéification ont été réalisés avec les charbons caractérisés. Des échantillons provenant de doublures réfractaires,

gaines et filtres à la sonde de gaz in-situ, l'embouchure du mâchefer, le pot de mâchefer, l'eau de décharge et l'absorbeur ont été recueillis et caractérisés. L'analyse de cendres volantes indique que la plupart de la matière inorganique est fondue et en forme de sphères. La dévolatilisation et l'interaction avec le réfractaire ont modifié la composition du mâchefer. L'effritement du réfractaire a été observé. Le mâchefer sur les doublures d'alumine est discontinu et forme des taches circulaires. Le mâchefer sur les doublures d'alumine-chrome est rugueux et bulleux. La fraction de pénétration par le mâchefer dans les doublures réfractaires a été déterminée.

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# Chapter 1. INTRODUCTION

## 1.1 Motivation to develop gasification

Worldwide, coal is used for roughly 39% of all electricity generation and 24% of total energy usage [1]. Not only is coal cheap, but it is also plentiful [2]. There however are many downsides to coal consumption. For example, conventional coal burning produces considerably more carbon dioxide per unit of electricity when compared to burning oil or natural gas [3]. One of the reasons for this is the low efficiency of coal energy conversion. Standard plants burn coal in a boiler to produce heat. This heat is used to produce steam which will turn a steam turbine. The mechanical energy from the turbine is then converted to electricity. Integrated gasification combined cycles (IGCCs) provide an alternative to generate electricity from coal. They allow for thorough gas clean-up and the technology is a front-runner for carbon capture and sequestration projects. From a Canadian perspective, front-end engineering and design (FEED) work was conducted on a 270 MW coal-fired IGCC that may be constructed in Alberta [4]. Capital Power announced in October 2009 that it would not move forward with construction of this plant at present due to the drop in electricity price, although the FEED will be completed and could serve as a blueprint for a future project. Alternatively, gasification may prosper in Alberta due to a suggested scheme which integrates petroleum coke gasification with carbon capture and storage to yield hydrogen, power, and steam production for oil sands upgrading with near zero CO<sub>2</sub> emissions. Notwithstanding, smaller scale gasification projects are already operating across Canada. Around the world, many gasification plants, including IGCCs, have been running for decades and new ones are continually being added to the commercial pipeline [5].

## 1.2 Fuels of interest

The fuels considered in this thesis are coals and petroleum cokes due to their potential for large-scale gasification in Canada. Coals are generally classified as anthracite, bituminous, subbituminous or lignite. Table 1.1 provides typical properties of each class. Canada's 6.6 billion tonnes of recoverable coal reserves contain all four classes of coal [7]. As for petroleum cokes, Furimsky estimated that the petroleum coke produced by Syncrude and Suncor in Alberta could generate 1000 MW if used in an IGCC plant [8]. Typical properties of the petroleum cokes are provided in Table 1.2. The composition of the fuel ash must be considered when studying slagging in gasifiers. Typical composition ranges of coal ash major and minor oxides are given in Table 1.3. Typical composition ranges of Syncrude and Suncor petroleum coke ash major and minor oxides are given in Table 1.4.

**Table 1.1: Properties of coals (adapted from [6])**

	<b>Anthracite</b>	<b>Bituminous</b>	<b>Subbituminous</b>	<b>Lignite</b>
Moisture (%)	3-6	2-15	10-25	25-45
Volatile matter (%)	2-12	15-45	28-45	24-32
Fixed carbon (%)	75-85	50-70	30-57	25-30
Ash (%)	4-15	4-15	3-10	3-15
Sulphur (%)	0.5-2.5	0.5-6	0.3-1.5	0.3-2.5
Hydrogen (%)	1.5-3.5	4.5-6	5.5-6.5	6-7.5
Carbon (%)	75-85	65-80	55-70	35-45
Nitrogen (%)	0.5-1	0.5-2.5	0.8-1.5	0.6-1.0
Oxygen (%)	5.5-9	4.5-10	15-30	38-48
MJ/kg	27.9-31.4	27.9-33.7	17.4-23.3	14.0-17.4

**Table 1.2: Properties of petroleum cokes (adapted from [8])**

	<b>Syncrude</b>	<b>Suncor</b>
Moisture (%)	0.4-0.7	-
Volatile matter (%)	4.8-6.3	12-13
Fixed carbon (%)	85-90	83-85
Ash (%)	4.8-7.5	3.0-4.0
Sulphur (%)	6.1-6.9	5.7-6.0
Hydrogen (%)	1.5-1.8	3.6-3.9
Carbon (%)	80-84	83-85
Nitrogen (%)	1.7-2.0	1.3-1.8
Oxygen (%)	0.9-2.0	0.7-1.3
MJ/kg	~32	~35

**Table 1.3: Coal ash major and minor oxides expressed as wt.% (adapted from [5])**

SiO <sub>2</sub>	28-62
Al <sub>2</sub> O <sub>3</sub>	11-30
TiO <sub>2</sub>	0-2
Fe <sub>2</sub> O <sub>3</sub>	4-30
CaO	0-28
MgO	0-5
Na <sub>2</sub> O	0-3
K <sub>2</sub> O	0-2
SO <sub>3</sub>	1-15
P <sub>2</sub> O <sub>5</sub>	0-3

**Table 1.4: Petroleum coke ash major and minor oxides expressed as wt.% (adapted from [8])**

	<b>Syncrude</b>	<b>Suncor</b>
SiO <sub>2</sub>	37-51	37-43
Al <sub>2</sub> O <sub>3</sub>	20-25	21-23
TiO <sub>2</sub>	2.8-4.9	3.2-3.4
Fe <sub>2</sub> O <sub>3</sub>	8-12	11-12
CaO	1.6-4.3	3.4-5.0
MgO	1.2-1.7	1.5-3.0
Na <sub>2</sub> O	1.1-1.7	0.6-0.8
K <sub>2</sub> O	1.7-2.0	0.9-2.0
SO <sub>3</sub>	1.8-3.6	2.5-5.4
P <sub>2</sub> O <sub>5</sub>	0.2-0.4	~0.3
V <sub>2</sub> O <sub>5</sub>	3.2-5.0	~4.4
NiO	-	1.2-1.3

### 1.3 Gasification basics

By definition, gasification is the conversion of any carbonaceous fuel to a gaseous product with a useable heating value [5]. Combustion is not a form of gasification because the flue gases produced have no residual heating value. However, pyrolysis, partial oxidation and hydrogenation are all forms of gasification. In the context of IGCCs, the gasification term implies partial oxidation by pure oxygen or air. Carbonaceous fuels currently used for gasification include coal, petroleum coke, natural gas, biomass and municipal waste. Syngas uses are numerous and include the formation of chemical products (e.g., methanol and ammonia), liquid fuel production via Fischer-Tropsch synthesis, hydrogen production, as well as electricity generation. A single gasification plant may consume various fuels while aiming to use syngas for a variety of applications. Many types of gasifiers exist, such as moving-bed, fluid-bed and entrained-flow types. However, the majority of the successful coal gasification processes developed after 1950 are entrained-flow slagging gasifiers operating at pressures of 2-7 MPa and temperatures exceeding 1400 °C [5]. Figure 1.1 shows a schematic diagram of a top-fired entrained-

flow slagging gasifier. Generally, fuel is fed to the gasifier with oxygen, air and/or steam. The fuel ash forms a relatively inert silicate melt (i.e., slag) which flows out the bottom of the gasifier. The product stream is synthesis gas, also called syngas, which is mainly rich in hydrogen and carbon monoxide, but also contains carbon dioxide, methane and steam [9]. Additionally, particulate matter (fly ash and unconverted char) and undesired sulphur and nitrogen compounds may be present in the syngas.

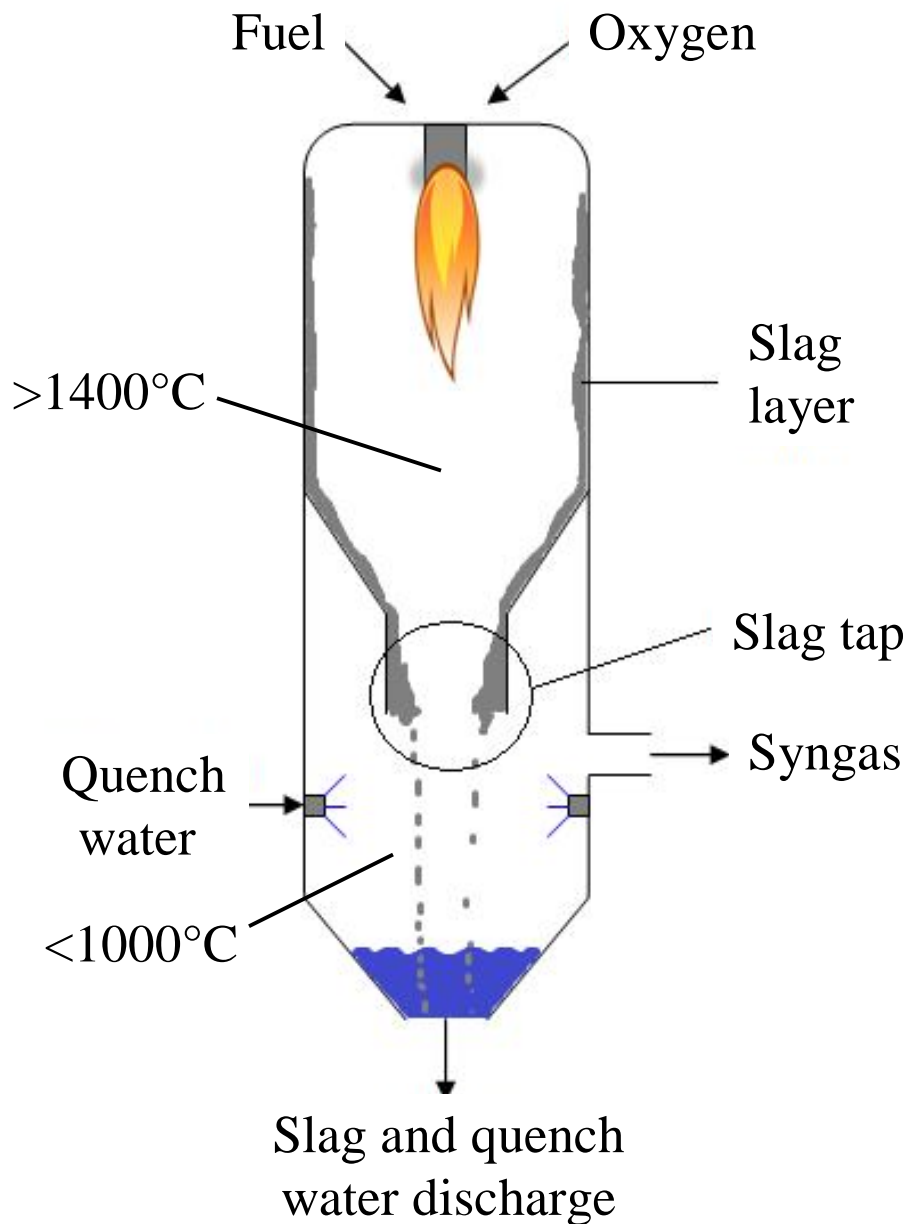
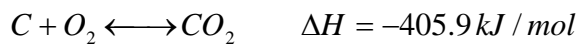


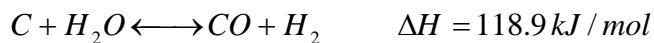
Figure 1.1: Schematic diagram of top-fired entrained-flow slagging gasifier.

Coal and petroleum coke particles can be considered as a grouping of moisture, organic volatile matter, fixed carbon and inorganic matter. The fuel particles are injected into the gasifier with O<sub>2</sub> or O<sub>2</sub>-enriched gases in an intense mixing field that can exceed 2000 °C [10]. Heating rates can approach 10<sup>5</sup> °C/s. According to Liu and Niksa [11], there are three main steps for the conversion of the particles during gasification. The first step is devolatilization. Devolatilization commences between 400 and 600 °C and can take as little as several milliseconds to attain completion. The moisture quickly vapourizes while the particles emit volatile substances leaving behind char. The released volatiles will further react to become ultimate products (i.e., components of syngas). During devolatilization, fuel particles can lose as little as a few percent or as much as 80% of their weight [12]. The second step of fuel conversion is char oxidation. Although there is not enough O<sub>2</sub> to fully oxidize the char, any O<sub>2</sub> remaining after devolatilization will react with the char since O<sub>2</sub> is the most effective agent for gasification [10]. The last step is char gasification by reactions with CO<sub>2</sub>, H<sub>2</sub>O, CO and H<sub>2</sub>.

Char oxidation and partial oxidation by O<sub>2</sub> follow these reactions [13]:

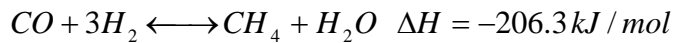
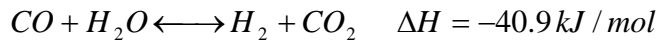


Note that the enthalpies are for reaction at 25 °C and 101 kPa. These reactions provide heat which will dry the coal, break chemical bonds in the coal, raise the temperature of compounds and drive the following char gasification reactions:





Moreover, equilibrium of the following gas phase reactions will dictate the product gas composition:



It should be noted that in entrained-flow gasifiers, some slag build-up on the reactor's inner surface is desired to protect refractory lining and reduce heat loss. The slag will flow down the walls and be removed from the bottom. However, a great concern in gasification design is plugging by slag, with an often quoted maximum slag viscosity of 25 Pa·s at the slag tap for proper operation [14,15]. This viscosity can be compared to the viscosity of common fluids presented in Table 1.5. To ensure proper slag flow, one option is to raise the gasifier operating temperature. This can significantly increase oxygen demand and reduce the process efficiency [5]. Another option is to blend fuels or add a fluxing agent. Possible fluxing agents include limestone and dolomite due to their availability, low cost and elevated content in calcium and magnesium (which generally reduce viscosity). Viscosity predictions for typical gasifier conditions and slag compositions can be used for initial feasibility assessments, while advanced gasification computational fluid dynamics models require local slag viscosity predictions to estimate sticking probabilities and flow throughout the reactor [17-19].

**Table 1.5: Viscosity of common fluids (adapted from [16])**

Fluid	Temperature (°C)	Viscosity (Pa·s)
Water	20	0.001
Honey	40	1.5
Mayonnaise	25	5
Chocolate	50	17
Cheese	70	30

## 1.4 Research objectives and outline

The ultimate goal of this doctoral research program is the optimization of the design and operation of gasifiers, particularly entrained-flow gasifiers running on Canadian fuels. Of the many aspects involved in gasification, slagging is of great interest since it can drastically alter operation and even completely halt it. Unfortunately, slagging behaviour is still ill-understood due to modeling difficulties and limited data. As a result, the three objectives advanced by this doctoral thesis specifically consider data collection and modeling related to slagging in order to improve the accuracy of already developed and soon-to-be developed comprehensive gasifier models.

The first objective is to develop slag viscosity models which are applicable to Canadian fuels. Many models are available but their predictive power is poor unless they are applied to specific fuels and conditions. Work has been done to improve these models and new approaches, such as the application of artificial neural networks and thermodynamic equilibrium, have been tested. Still, all slag viscosity models are empirical or semi-empirical and therefore rely on slag viscosity measurements. For a model to appropriately describe the viscosity of a given slag, it is imperative that such a slag's composition and temperature fall within the range of compositions and temperatures used to produce the model. Although many viscosity values are available for many slag compositions and temperatures with various fluxing agents, few values are available for slags resulting from gasification of Canadian fuels and fluxing agents. Hence the second objective is to perform slag viscosity measurements pertinent to Canadian gasification fuels and fluxing agents. In particular, the effect of vanadium in petroleum coke is examined. The third objective is to collect pilot plant data which will lead to a better understanding of slagging while also providing data which can be used for gasifier model validation.

Chapter 2 is a literature review of inorganic matter transformation sub-models for entrained-flow slagging gasifiers. This chapter highlights the complexities of modeling slag formation and flow in gasifiers. Sub-models for ash particle formation, gas-particle

transport, particle sticking, slag flow and slag-refractory interactions are presented and discussed. The importance of slag viscosity should be apparent after reading this chapter. Chapter 3 describes the development of artificial neural networks to model slag viscosities over a broad range of temperatures and compositions. Chapter 4 presents tools developed to assess various slag viscosity models. These tools include a slag viscosity prediction calculator with 24 slag viscosity models, and a database of 4124 slag viscosity measurements. Chapter 5 describes results of a slag viscosity measurement program for slags produced from coal and petroleum coke blends. Additional details of the viscosity measurement setup assembled at CanmetENERGY for this program are provided in Appendix A. Chapter 6 describes interactions between petroleum coke slags and various crucible materials used for slag viscosity measurements. Chapter 7 is a Canadian fuel characterization study specifically designed to provide information required for modeling inorganic matter transformations in gasifiers. Chapter 8 presents results from five pilot scale gasification tests performed with the fuels characterized in the previous chapter. The final chapter recaps the major conclusions and includes a discussion on recommended future work to be conducted.

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# Chapter 2. Literature review of inorganic matter transformation sub-models for entrained-flow slagging gasifiers

## 2.1 Introduction

After volatilization, char combustion and char gasification, what is left of the coal particles is mostly inorganic matter. Mineral classes present in coal include (in general order of abundance) silicates, carbonates, oxyhydroxides, sulphides, sulphates, phosphates and others [1]. Descriptions, images and chemical characteristics of minerals are available online [2]. The fate of the inorganic matter in a gasifier can take one of two directions. The first is entrainment by the gas phase where the inorganic matter leaves the gasifier as fly ash. In this case, it must be removed by some particle removal process prior to downstream processing of the gas. Alternatively, the fly ash may stick to the gasifier walls, flow down and exit the bottom as a highly viscous by-product named slag. This process has many implications as the slag can greatly affect heat transfer in the system, corrode/protect the refractory lining of the gasifier, and even plug the gasifier thus halting its operation. Hence there is much interest in predicting how the inorganic matter will behave when designing and operating a gasifier.

Many qualitative indices have been devised to quickly and easily predict ash slagging/fouling in combustion and gasification systems. Some are simply based on X-ray fluorescence ash analysis (ASTM D 3174), such as the base-to acid ratio [3]:

$$\frac{Base}{Acid} = \frac{Na_2O + K_2O + CaO + MgO + FeO}{SiO_2 + Al_2O_3 + Fe_2O_3 + TiO_2}$$

where chemical formulas denote molar fractions. A low ratio hints towards a higher melting point and viscosity; hence a greater likelihood of gasifier plugging. Some indices are based on more complex coal ash analyses such as ash fusion temperatures (AFTs) (ASTM D 1857) or even computer controlled scanning electron microscopy (CCSEM) data. Gupta [4] and van Alphen [5] provide good reviews of various coal characterization techniques including AFTs and CCSEM. The University of North Dakota Energy and Environmental Research Centre (UNDEERC) analysed bench, pilot and industrial scale data to develop indices for slagging and fouling in power plants [6]. However, slagging/fouling indices do not account for the multiple chemical and physical phenomena coal ash experiences. As a result, they may provide misleading results and are often limited to specific fuels and reactor designs [7,8].

Hence to obtain reliable quantitative predictions for inorganic matter transformations and interactions within a gasifier, various inorganic matter phenomena must be studied in detail with models describing each. In this review, inorganic matter phenomena in gasifiers are grouped as follows; ash particle formation, gas-particle transport, particle sticking, slag flow and slag-refractory interactions. Note that much of the available data and models are from/for combustion systems, but are still applicable (sometimes with slight modifications) to gasification.

## **2.2 Ash particle formation**

As the water, volatiles and carbon are removed from a coal particle, the resulting ash inorganic matter may coalesce, remain separate or even fragment. Ash particles produced by combustion or gasification of a given coal display an array of sizes and compositions [9]. Bulk coal ash analyses are inadequate indicators for ash particle formation since differences in individual particle size and composition have a strong influence on particle behaviour and must be accounted for [8]. For instance, ash is formed by inorganic material in coal which can be associated to the carbon matrix in the form of minerals

embedded in the carbon matrix (included minerals), or in the form of minerals outside of the carbon matrix (excluded minerals) [9-12]. Inorganic material may also be organically-bound in the carbon matrix. These are mainly alkali and alkaline earth metals. They are often grouped with included minerals or neglected, particularly with high rank coals which tend to have little organically associated inorganics. The manner in which an inorganic component is present in coal has a huge impact on its fate in gasification. The dimensions of the minerals and coal particles also have a considerable impact. It is impossible to characterize and track every single particle entering the gasifier. Therefore, size, composition and association distributions must be determined in a statistically meaningful manner. Two techniques used to determine these distributions are chemical fractionation and CCSEM. Following a discussion on these techniques, models describing ash particle formation are discussed.

### **2.2.1 Chemical fractionation and CCSEM**

Chemical fractionation is a technique that provides information on the association of inorganic species in coal using selective extraction based on solubility [4,13]. The process consists of three successive extractions: (1) using water to remove water-soluble salts; (2) using ammonium acetate to remove ion exchangeable elements; and (3) using hydrochloric acid to remove acid-soluble species. Chemical fractionation is suitable for quantification of type and abundance of organically associated inorganic species, particularly in lower-rank subbituminous and lignitic coals which contain greater amounts of these [9]. Doshi *et al.* [11] modified the chemical fractionation technique for biomass fuels which also contain many organically bound inorganic species.

CCSEM can provide distribution of coal particle size, coal particle type, mineral particle size, mineral particle type and mineral particle association (i.e., included or excluded) [4,5]. Some CCSEM systems can also detect, characterize and quantify organically bound inorganics. It should be noted that in this discussion, “CCSEM” refers to techniques which, in an automated fashion, utilize electron microscopy and X-ray signal detection (i.e., CCSEM, QEMSCAN and MLA). A typical CCSEM analysis is as follows. Coal

particles, prepared as they would be for feeding into a combustor or gasifier, are mixed with an epoxy mounting medium and the analysis is done on a polished cross-section of the coal-epoxy plug. Back-scattered electron intensity is used to discriminate between mineral particles, coal macerals and epoxy background. Once a mineral particle is located by SEM image intensity, the cross-sectional area of the particle is determined by extending eight segments from the centre to the edge, and summing the areas of triangles delimited by these segments. Coal particles can be located and sized by the same procedure. To determine the elemental composition of a mineral, an energy dispersive X-ray (EDX) spectrum is collected either at a single point on the mineral, or at multiple points to get an average. Coal particle size/location, mineral particle size/location and mineral composition are stored. Mineral particles located within a coal particle are deemed included, while others are deemed excluded. Several thousand coal and mineral particles are analyzed in this manner to provide statistically significant results.

Caution must be applied when using data from a CCSEM analysis. There are known inherent errors with this technique based on the fact that a three-dimensional structure (coal and mineral particles) are analyzed by two-dimensional sampling (cross-section). Stereology is entirely dedicated to this problem and it presents itself in many fields [14]. For example, the volume of a particle cannot be determined from the area of a random cross-section. Also, the number of profiles per unit area in two dimensions does not equal the number of objects per unit volume in three dimensions. Mathematical manipulation of CCSEM data to correct for this is possible [15,16]. However, equations involved may be very complex and require assumptions which may not have been verified. An approach which randomly distributes included minerals in coal particles has been proposed [7,17,18]. In this approach, once coal particles (with area extrapolated to volume) and included minerals (with composition determined and area extrapolated to volume) are analyzed by CCSEM, representative simulated mineral-free coal particles are created. Included minerals are distributed, one by one, into the randomly selected simulated coal particles. The probability of selecting a coal particle depends on the particle volume fraction in the total volume of all the coal particles. If the mineral-free volume of the selected coal particle is greater than the mineral volume, then the mineral is added to the

coal particle and the mineral-free volume is adjusted. Otherwise, another coal particle is selected. This continues until all included minerals have been placed in the simulated coal particles. It is unclear whether the selection of a simulated coal particle is based on its total volume or mineral-free volume. It may be more logical to base the selection on the mineral-free volume. It is also unclear what is done if a mineral does not fit in any mineral-free coal volume. This random attribution algorithm does require some assumptions and does not correct all stereological errors. Other techniques may involve sieving the coal to obtain a size distribution (although this also requires many unverified assumptions). Laser diffraction may also be used to determine the particle size distribution [19]. It may be necessary to also separate the particles by density if many excluded minerals are present. Clearly, there is room for improvement in mineral and coal characterization. Recent work applied computer tomography to determine the three-dimensional shape of coal particles as well as the shape of included minerals [20]. This and other new techniques will be required to obtain true size, composition and association distributions.

### **2.2.2 Ash formation**

When modeling the formation of ash, the mechanisms affecting included, excluded and organically associated inorganics are quite different. First, for included inorganics, two extreme ash particle formation models exist. The first is “full coalescence” which assumes that all included minerals in a char particle coalesce together to form one ash particle. The second is “no coalescence” which assumes that all included minerals in a char particle remain separate and form separate ash particles. Unfortunately, for many cases, neither extreme is applicable. Hence a partial coalescence model must be applied [7,21]. The degree of coalescence is dependent upon the coal composition, coal particle size distribution, the mineral matter distribution, char burnout mechanisms and char fragmentation [22]. Different coal ranks with different macerals will form various types of char particles [23]. During volatilization, softening and/or swelling of coal, combined gases escaping will lead to the formation of pores and possibly an overall cenospheric shape with a given wall thickness [18]. When it comes to modeling the formation of ash

particles from included inorganics, the following several basic assumptions are usually made regarding the char structure and included mineral behaviour [16,18]: (1) The char particles are spherical, either solid spheres or cenospheres with a uniform wall thickness. (2) The char particles are consumed uniformly with the oxidation reaction only occurring on the outer char surface, causing the outer surface to recede. (3) The included inorganics are fully molten spheres which are randomly distributed throughout the char structure with uniform probability based on char volume. Burning char is typically at a temperature well above the melting point of typical minerals [22]. (4) Surface tension forces are sufficient to keep the inorganic spheres on the char surface without movement. (5) There are no miscibility limits or viscous flow resistances in coalescence.

As the char surface recedes, included inorganics on the outer surface of the char can come into contact and coalesce with particles nearby on the surface, or beneath them within the char. In the case of a cenospheric char particle, coalescence can proceed till the centre void of the char particle is reached and the inorganic spheres are released. In the case of a solid char particle, coalescence can proceed till all the char is reacted or completely enveloped by inorganics (which results in one ash particle, agglomerating all the included inorganics from that char particle). However, with both cenospheric and solid char particles, the inorganic particles can be released during recession if they encounter a char pore large enough to separate them from the char structure [18,22]. Char pores range in size from Angstroms to tens of microns. However, only macropores (>1 micron) are of interest for ash release. To simplify the ash particle formation model, it is possible to consider “ash separation length,” which is the average distance an inorganic particle on the outer char surface recedes until it is separated from the char, either by encountering a pore or the centre of the cenosphere [18,22]. Hence, in the ash particle formation model, all char particles are assumed to be cenospheric without pores, and are attributed an apparent wall thickness which represents the ash separation length. The apparent wall thickness can be expressed as a percentage of the char particle diameter. A char particle with high macroporosity and/or thin wall has a low apparent wall thickness percentage. A solid char particle with low macroporosity has a high apparent wall thickness percentage.

The char particle apparent wall thickness is generally an input to the ash particle formation model. It may be determined through experimentation and/or based on coal and char characterization. Char particles are often classified within one of three groups [18,23]. Char Group I particles are thin-walled, cenospherical and contain a continuous large cavity. Char Group III particles are solid with little or no void. Char Group II particles have intermediate characteristics, with discontinuous voids of various sizes. Yan *et al.* [18] associated apparent wall thicknesses of 5, 20 and 40% to Char Groups I, II and III, respectively.

A simple ash formation model for included inorganics can be carried out as follows [16,18]. Once simulated coal particles with included minerals are produced (which is likely based on statistical data from CCSEM), the type of char (i.e., Char Group I, II or III) particle formed is determined. Then based on the apparent char wall thickness (which is related to the Char Group), inorganic particle sizes, coal particle sizes and inorganic volume fractions in the coal, the probability of coalescence can be calculated. Monroe [16] calculated the percent of included inorganics that coalesce, assuming various mono-size and mono-density coal particles with mono-sized inorganic particles. These calculations were done for various apparent char wall thicknesses. Yan *et al.* [18] related these results to the char groups. Some other basic assumptions can be applied to the inorganic matter, regardless if it is included or excluded; pyrite will release sulphur, clays will release water and carbonates will release carbon dioxide [9,16]. Furthermore, ash particles formed by included mineral coalescence will also include many organically bound inorganics, although some of these organics may devolatilize and join the gas phase [9,15,22]. Distribution of non-devolatilized organically bound inorganics added to coalesced included minerals can be assumed proportional to the surface area of the ash particles.

In literature, there are some variations to the above assumptions and model for included inorganics. Barta *et al.* [15] uses a combustion model with shrinking core, down to a certain radius (dependant on particle reactivity), after which the radius remains constant,

porosity increases and the char disintegrates. Instead of relating the ash particles to coal particles, Baxter [24] characterized char particles to then model ash particle formation. Lee and Lockwood [25] simplified their ash transformation model further by sampling and characterizing fly ash directly.

Modeling the formation of ash particles from excluded inorganics is simpler than with included inorganics. Excluded inorganics can either stay intact to form one ash particle each, or fragment to form multiple ash particles. Fragmentation can be caused by thermal shock at mineral surfaces and rapid gas formation and release from inside the inorganic particles [18]. Benson *et al.* [9] mention that their model applies experimentally determined fragmentation values for pyrite, pyrrhotite, oxidized pyrrhotite, Ca-Al-P, kaolinite, montmorillonite, gypsum, barite, calcite, dolomite and iron carbonate. Yan *et al.* [18] use a Poisson distribution to describe the fragmentation of excluded minerals. It was assumed that all particle offsprings have the same size. Fragmentation was neglected for some major mineral species (quartz, illite, kaolinite, montmorillonite, siderite, ankerite). Collisions between coal/char particles, excluded inorganics and ash particles were neglected. However, this assumption may only be applicable to certain reactor types and conditions.

Ash particle formation models have been compared to experimental data from various combustion facilities [7,9,15,16,18,24]. In some cases, simple no-coalescence or full-coalescence models compared well. In other cases, partial coalescence was required to properly describe results. However, in some studies, even partial coalescence failed to provide proper predictions. Some models were only good for ash particle size predictions or ash composition predictions, but not both. It is important to note that all these comparisons with experimental data are for combustion conditions. Very little literature covers ash formation specifically in gasification conditions, hence the applicability of the models to gasification may be doubtful. Some insight may be given by work done by Li *et al.* [12]. They collected char particles at various stages of conversion which were generated in a laminar entrained-flow reactor with temperatures and a gas atmosphere simulating gasifier conditions. SEM images of particles generated at 1400 °C showed the

transformation of the mineral-carbon association. Minerals were melted throughout most of the conversion process (indicated by droplet-shaped bright particles). Most minerals at the char surface seemed to remain attached as there were not many liberated ash particles observed when char was still present. As carbon was further consumed, more molten inorganic droplets are present at the char surface. The droplets at the surface gained in size at higher char conversion, and very large particles covered by inorganics could be observed at very high carbon conversions (>90%). These observations suggest that many of the assumptions in ash formation models are valid with gasification conditions. Nonetheless, much more data with other types of coal are necessary.

### **2.3 Gas-particle transport**

Transport of particles entrained by gas is well documented for various processes. For the most part, the general concepts can be applied to the transport of coal, char and ash particles in an entrained-flow gasifier. Hence, only a very basic overview of general concepts is given here. More focus will be given to transport concepts for particles in a gasifier (or combustor) near a reactor surface as this has a profound impact on the fate of inorganic matter.

For gas-phase fluid mechanics, it is usually assumed that variables such as velocity, pressure and density are continuous point functions. This is known as the continuum hypothesis [26]. Of course continuums do not exist on a microscopic scale since mass is concentrated at atomic nuclei. However, on a macroscopic scale, this hypothesis usually holds and mass and momentum can be considered to be spread uniformly throughout a given volume. The same cannot be said of particle or droplet phases. Conservation equations of Eulerian form apply to continuous phases. On the other hand, Lagrangian equations follow the trajectory of each particle or droplet individually, making it impractical for continuous phases. A purely Eulerian framework can be used to describe gas-particle flows. Alternatively, a Eulerian gas-phase description can be combined with

a Lagrangian particle phase description. Although fundamental laws usually apply to a Lagrangian framework, the Reynolds transport theorem can be used to convert them to a Eulerian framework. Smoot and Smith [26] provide a more thorough overview of this subject.

One of the greatest challenges in describing flow inside a gasifier is to accommodate turbulence. The problem is compounded by the presence of multiple phases. For practical considerations, conservation equations must be modified to account for turbulence. The  $k$ - $\epsilon$  two-equation closure model is often used in combustion and gasification models. Lee and Lockwood [25] and Watanabe and Otaka [27] describe the flow dynamics and particle transport equations used in their CFD models.

Once particles are brought near a surface in the reactor, they must deviate from the gas flow field to come into contact with the wall. Mechanisms like inertial impaction, thermophoresis and eddy impaction can provide the energy required to overcome the flow field [8,13,25,28,29]. When contact with the surface is established, other forces will determine whether the particle sticks to the surface or rebounds (see Section 2.4). Surface deposition may also be directly from the gas phase, by condensation or chemical reaction. Inertial impaction (including some turbulence effects) and thermophoresis are briefly described below. Other means of surface contacting are not as well understood and could be less significant. They are discussed in work by Baxter [13].

For larger particles ( $>1 \mu\text{m}$ ), inertial impaction is the dominant mechanism [8,25]. Momentum normal to the surface, gained by particles from gas entrainment away from the surface or from turbulence, must overcome the gas stream lines near the surface. The Stokes number has been correlated to impaction efficiency [30]. Baxter [13] has demonstrated that for ash particles flowing towards a boiler heat transfer tube in typical combustion conditions, the effects of the laminar boundary layer around the tube, thermophoresis and turbulence are negligible.

Effects of the laminar boundary layer and turbulence may not be neglected for inertial impaction on flat plates in parallel flow [13]. Correlations relating impaction efficiency to the Stokes number are available, but their validity is less certain than for inertial impaction in cross-flow [13]. Note that in some instances, swirled tangential flow may be promoted in the gasifier chamber to increase particle-wall collisions [29].

Thermophoresis causes particle transport due to local temperature gradients within the gas and/or the particle. In general, the thermophoretic force acts in the direction opposite to the temperature gradient. With cross flow over a tube, deposits from thermophoresis are more evenly distributed and finer grained than inertial impaction deposits. Also, as an insulating layer of particles builds up on the surface, the temperature gradient decreases leading to less thermophoretic deposition [30]. Correlations describing thermophoresis are given by Baxter [13]. However, Baxter states that the accuracy of these is questionable.

In their combustion model, Yan *et al.* [21] consider inertial impaction for larger particles. They consider Brownian diffusion and thermophoresis for finer particles. They divide the gas flow field into three zones; the fully turbulent core, the buffer layer and the boundary layer. Inertial impaction acts in all three zones, while Brownian diffusion and thermophoresis only act in the boundary layer due to insufficient temperature gradients in the other zones. Analogous to electrical resistances, the overall resistance of ash transport is calculated by summing individual resistances. Details of the individual resistances are given by Im and Chung [31].

## **2.4 Particle sticking**

Several factors may affect particle sticking. These include particle/surface temperature and composition (which relate to viscosity and interfacial tension), impaction angle, impaction velocity and particle size.

One approach to determine the sticking probability calculates the kinetic energy of the impacting particle and the energy dissipation at impact. If the particle has more kinetic energy than the amount which can be dissipated by the collision, it is assumed that the particle bounces off. Otherwise, it is assumed that the particle sticks. This approach is not widely adopted as it is difficult to predict the amount of energy which can be dissipated [30]. Mueller *et al.* [32] calculate the excess energy after impaction and assume the particle sticks if the energy is null or negative. For this, they applied the dimensionless Weber number (which compares kinetic energy of the particles and the interfacial tension energy between the particles and the substrate) and Reynolds number. If the excess energy is greater than zero, it is assumed that the particle rebounds. Surface tension of the particle and contact angle between the particle and substrate are required for this model. These values are not well defined for particles in combustion and gasification systems. Yong [33] simply assumed sticking when both the particle and surface are liquid, and rebounding when both are solid. When either the particle is solid and the surface is liquid, or the particle is liquid and the surface is solid, the Weber number is applied. Rebounding was predicted when this number exceeded 1. Otherwise sticking was predicted. However, Yong states that the effect of the Weber number and interference of trapped particles for further sticking should be studied more thoroughly. Ni *et al.* [34] used the Weber number, Reynolds number and contact angle to calculate the excess rebound energy. A particle is assumed to rebound if this energy is greater than zero, and it is assumed to stick otherwise. Sensitivity analyses were done for surface tension, viscosity, contact angle, impact angle and impact velocity.

Another approach to predicting sticking probability is to use a viscosity criterion. The extent to which a liquid droplet deforms upon impact depends on its kinetic energy and viscosity. If the particle and surface have similar surface energies per unit area (i.e., a small equilibrium contact angle), a larger contact area results in a greater sticking probability. A large contact area will be produced if the liquid particle's viscosity is low. Walsh *et al.*'s [28] sticking model is often applied, where the particle sticking probability is given by:

$$P_p = \frac{\eta_{ref}}{\eta_p}, \eta > \eta_{ref}$$

$$= 1, \eta \leq \eta_{ref}$$

where  $\eta_p$  is the particle viscosity, and  $\eta_{ref}$  is a reference viscosity below which the sticking probability is unity. Note that computational fluid dynamics (CFD) modellers often use the term *critical viscosity* instead of *reference viscosity*. This should be avoided as *critical viscosity* has a distinct meaning for slag viscosity modellers (i.e., viscosity at the temperature threshold below which viscosity commences to be affected by the presence of crystals [3]). In Walsh *et al.*'s model, the stickiness of the impacted surface ( $P_s$ ) is also considered using the above equation, with the particle viscosity replaced by the surface viscosity ( $\eta_s$ ). The net mass fraction of impacting particles depositing ( $f_{dep}$ ) is given by:

$$f_{dep} = P_p + [1 - P_p]P_s - k_e [1 - P_p][1 - P_s]$$

where  $k_e$  is an erosion factor, defined as the mass of ash removed per unit mass of non-sticking ash impacting. Erosion is only likely with particles sized above several millimetres [35] and is usually neglected. There is no clear consensus on the reference viscosity value. Values used range from 8 to  $10^8$  Pa·s [8,10,21,25,28,30]. Many empirical or semi-empirical slag (molten ash) viscosity models are available, relating viscosity to temperature and composition. However, predictions from different models may differ by orders of magnitude and it is difficult to select an appropriate model for a given temperature and composition. Also, it may be difficult to predict the particle temperature as it impacts the surface. The bulk gas temperature may be a suitable approximation. However, the particle temperature can be affected by char reactions, radiation and possibly the thermal boundary layer of the impacted surface. Also, when applying Walsh *et al.*'s [28] model, it is usually assumed that the particle is composed of homogeneous molten ash. This may not be representative of the real fly ash composition. Alternatively,

Richter *et al.* [8] determined the viscosity of the six major mineral categories in the fly ash they studied. The viscosity used to calculate the sticking probability for the fly ash particles was based on a weight-averaged viscosity of the minerals.

For both the kinetic energy dissipation approach and the viscosity approach, small particles are assumed to have a 100% sticking probability. One reason for this is that the kinetic energy which must be dissipated is much smaller. Another reason is that the types of forces bringing the particle to the surface may be different from inertial impaction. For their energy dissipation model, Mueller *et al.* [32] recommend neglecting rebounding with particles less than 75  $\mu\text{m}$ . Moreover, Yan *et al.* [21] and Rushdi *et al.* [10] assumed all particles less than 2-3  $\mu\text{m}$  stick with their viscosity-based sticking models.

The viscosity sticking criteria used in many studies is applicable to ash particles with no char remaining. However, if char is still present in the particle, the particle certainly will not have a homogeneous composition and viscosity cannot be calculated. Li *et al.* performed a series of experiments with partially converted char particles in a laminar entrained-flow reactor under gasification conditions [12,36]. It was observed that up to a certain char conversion (~90% for Illinois #6 coal), few particles were able to stick after impaction onto an inclined plate. Past this critical conversion, almost all particles were able to stick, provided they were at a temperature above the bulk ash flow temperature. This critical conversion was related to the char-slag transition by SEM and surface area analysis. The char-slag transition is when the porous non-sticky char transforms into non-porous sticky slag. The low particle sticking before reaching the transition conversion was attributed to excluded inorganics and liberated included inorganics. Hence a new sticking criteria was developed, based on char conversion of the particles. This critical conversion can be determined for a given coal by measuring the surface area of char particles with various degrees of conversion. A significant drop in surface area should be observed at and above the critical conversion.

Still, Li *et al.* used a bare plate for their sticking experiments. The char-slag transition conversion may be useful to predict sticking during the initial use of a slagging gasifier or

for cooled heat exchange surfaces, but once a layer of slag has accumulated on the reactor walls, it is important to consider the wall stickiness as well. For gasifiers which have developed a steady-state flow of low-viscosity slag on their walls, particles are likely to stick, regardless of their char content. A pure ash particle impacting molten slag is expected to dissolve and modify the slag composition. A char particle impacting slag will not necessarily dissolve in the slag. The fate of the char particle has a huge impact on gasifier efficiency, since a char particle which sticks to slag and stays at the surface will have increased residence time and greater conversion, while a char particle which submerges into the slag will have modified reactivity. Shannon *et al.* [37] performed a force balance on particles impinging on molten slag. They considered drag force, capillary force and fluid-added mass force. Depending on the conditions, the particle could settle on the surface without oscillation, settle on the surface with oscillation, be rejected by oscillation, or get entrapped. The key properties in the force balance are particle size and density, particle velocity (normal to the slag surface), slag viscosity and interfacial tension. In a sensitivity analysis, it was shown that the fluid added force plays a very minor role, and that the drag force is dominant. Montagnaro and Salatino [29] developed a simple “order-of-magnitude” criterion to determine whether an impacting particle would be entrapped by slag. For a particle to settle on the surface, its kinetic energy must be overcome by the cumulative work of interfacial and viscous forces. However, they show that the interfacial work is negligible for a typical char particle. The resulting criterion for settling to occur is:

$$du > \frac{36\mu}{\rho}$$

where  $d$  is the particle diameter,  $u$  is the impact velocity,  $\mu$  is the slag viscosity, and  $\rho$  is the particle density. Montagnaro and Salatino further hypothesize that in practical entrained-flow gasification, it is likely that a monolayer of char covers the slag surface, which prevents sticking by other impacting char particles. This would result in a dense-dispersed phase of gas-entrained char particles near the wall. This could increase the average particle residence time and overall carbon conversion. Experimental data for

char-slag interaction and effects on reactivity are emerging [38], but more needs to be done.

## 2.5 Slag flow

Once the rate of ash deposition is known, slag flow down the walls of the gasifier can also be characterized. The basic principles of slag flow are those of a falling film. Momentum conservation can describe the velocity profile. Mass conservation must be included to describe slag flowing in and out of a control volume, as well as the ash sticking to the wall or being re-entrained by the gas phase. Since significant temperature gradients will vary the slag properties, energy conservation must also be solved.

Consider the layer of slag built up on an inner gasifier wall in Figure 2.1. In this figure, there is a single refractory wall layer for demonstrative purposes, although the wall can be a combination of refractory materials, a water/gas cooling membrane and steel vessel layers. The wall inclination ( $\alpha$ ) is usually  $90^\circ$ , but can vary with different gasifier configurations or sections. The slag layer is commonly divided in a liquid slag layer and a solid slag layer. Here, these layers are more correctly termed immobile and flowing (as solids and liquids can be present in both). The immobile layer is composed of slag which is above a certain viscosity, rendering its mobility negligible. The temperature of critical viscosity is often used to distinguish between the immobile and flowing slag layers [33,39-41]. However, not all slags demonstrate a temperature of critical viscosity. It may be more appropriate to use a direct viscosity criteria to distinguish between the immobile and flowing layers (e.g., assume slag is immobile above 1000 Pa·s). Assuming the presence of this layer does not simplify the slag flow model, but may be useful if there is a significant change in slag properties (e.g., thermal conductivity) from the flowing layer to the immobile layer. Nonetheless, in certain instances, there may be a flowing layer without an immobile layer, or vice versa.

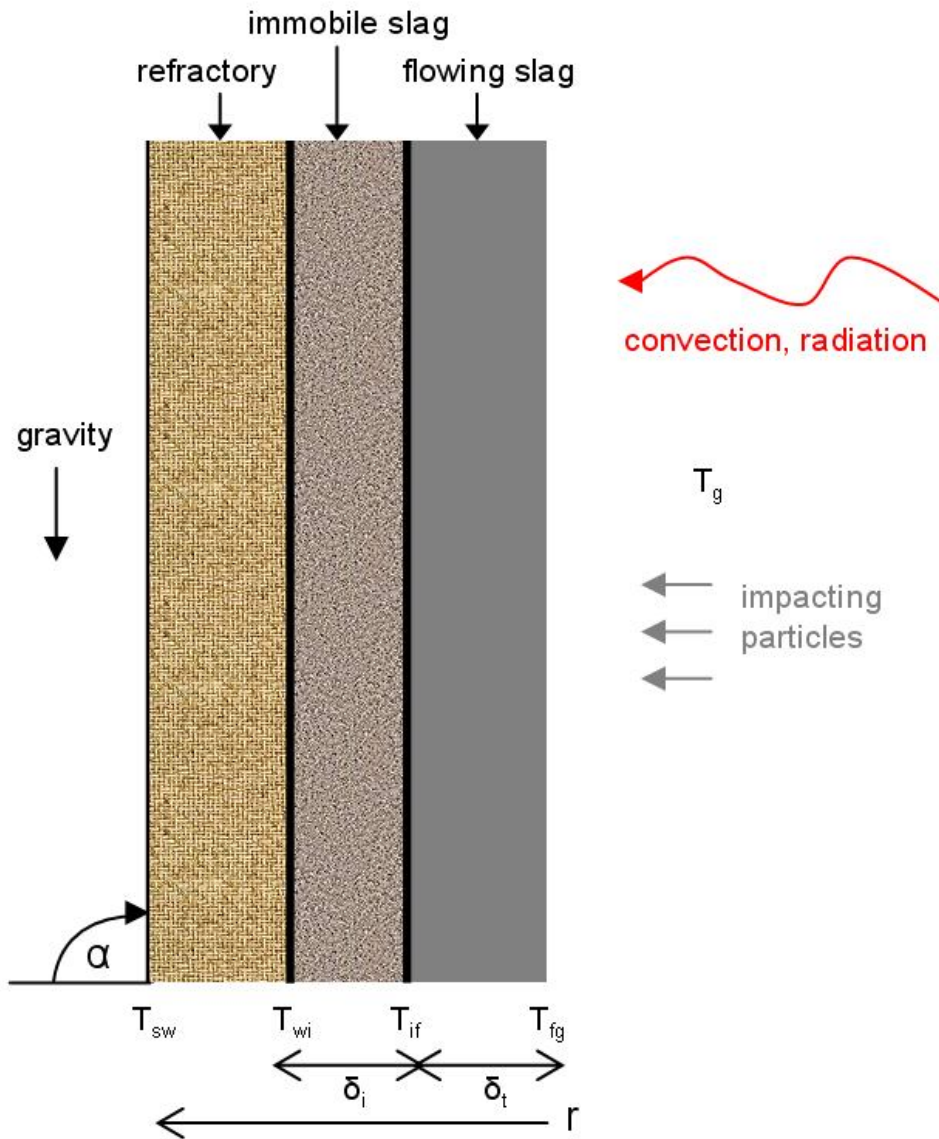


Figure 2.1: Schematic diagram of slag deposited on a gasifier wall.

Most slag flow models are based on the following momentum balance on a control volume of slag (in Cartesian coordinates):

$$\frac{\partial \tau}{\partial r} = -\rho g \sin \alpha$$

where  $r$  is the distance from the slag-gas interface,  $\tau$  is the shear stress,  $\rho$  is the slag density and  $g$  is gravitational acceleration. Shear stress is related to slag viscosity via Newton's law of viscosity:

$$\tau = \eta \frac{\partial u}{\partial r}$$

where  $\eta$  is the slag viscosity,  $u$  is the slag velocity. Integrating and combining the two equations above will give a velocity profile which is dependent on assumptions applied in the model.

Bockelie *et al.* [40] developed a steady-state gasifier slag flow model with a numerical solution. The model uses information from the gas flow field in the gasifier (gas composition, gas temperature, incident heat transfer, particle deposition rate) to predict slag properties (slag flow, slag thickness, immobile slag thickness) and heat transfer through the gasifier walls. The following assumptions are applied. (1) The transition temperature between the immobile and flowing slag layers is the temperature of critical viscosity. (2) The flowing slag has Newtonian behaviour. (3) The shear stress between the gas and slag layer is negligible. (4) The model is written in Cartesian coordinates with a constant heat flux through the slag. This assumption is valid if the slag layer is thin relative to the reactor diameter. (5) The heat transfer is normal to the surface. (7) The density, specific heat and thermal conductivity of the slag are independent of temperature.

Bockelie *et al.*'s model uses the following iterative procedure which starts from the top of the gasifier. First, the thickness of the slag layer ( $\delta_t = \delta_i + \delta_f$ ) is assumed. At the top of the gasifier, at  $y = 0$ , the slag layer thickness is defined to be zero. For  $y > 0$ , the slag layer thickness at the previous axial position is used as an initial assumption for the subsequent axial position. Then, using an iterative procedure based on radiation and conduction equations, the surface temperature of the slag layer and the net heat flux to the slag surface is calculated with:

$$\delta_t = \frac{k(T_{fg} - T_{wi})}{q_s}$$

where  $k$  is the slag's thermal conductivity.  $T_{fg}$  is the slag-gas interface temperature,  $T_{wi}$  is the wall-slag interface temperature, and  $q_s$  is the heat flux through the slag. The velocity profile (at a given height) in the flowing slag layer is given by:

$$u = \rho g \left[ \frac{k}{q_s} \right]^2 \int_{T_{if}}^T \frac{(T_{fg} - \theta)}{\eta} d\theta$$

where  $u$  is the local slag velocity,  $\rho$  is the slag density,  $T$  is the local temperature,  $\eta$  is the local slag viscosity, and  $\theta$  is a dummy variable.  $T_{if}$  is a predetermined temperature below which there is no slag flow. The mass flux of slag ( $\dot{M}_s$ ) at a given height is then expressed as:

$$\dot{M}_s = \frac{\rho k}{q_s} \int_{T_{if}}^{T_{fg}} u d\theta$$

Then, the thickness of immobile slag ( $\delta_i$ ) is calculated. The last step in the slag flow model is to compare the mass flux of flowing slag with the ash deposition flux from the gas. If these values agree within a reasonable tolerance (e.g., 0.1%), the assumed slag layer thickness is correct and the calculations can be repeated at a lower section of the gasifier until the bottom is reached. If the mass fluxes do not agree, then a new slag layer thickness should be assumed and the calculations repeated for the gasifier section in question.

Seggiani [39] developed a time-varying gasifier slag flow model with an analytical solution. The base assumptions are the same as in Bockelie *et al.*'s model. The gas temperature and concentration profiles are evaluated by a separate code. The slag flow model requires physical properties of the slag such as viscosity, density, specific heat and thermal conductivity. A specific viscosity equation form is used which allows analytical integration to obtain the velocity profile and mass flux. In the energy equation, the heat which leaves/enters with slag flow is included. Slag fusion heat is neglected. Seggiani's model is for gasifiers with a water cooling jacket. Pilli *et al.* [41] modified it for air-cooled gasifiers. Effectively this changes the heat transfer calculations.

Wang *et al.* [35] developed a steady-state slag flow model with an analytical solution for tilted cylindrical combustors. Assuming the slag viscosity is constant throughout the layer thickness, the velocity profile is given by:

$$u = \left( \frac{\rho g \sin \alpha}{3\eta} \right) \delta_f^2 + \frac{\tau_{fg}}{2\eta} \delta_f^2$$

where  $\tau_{fg}$  is the shear stress at the slag-gas interface. This shear stress is the combination of shear stress from the flowing gas and shear stress from the momentum of impacting particles. Wang *et al.* neglected the gas flow shear stress (although it may not be negligible in combustors or gasifiers with higher gas velocity). Shear stress from impacting particles was calculated from the kinetic energy of the particles.

Li *et al.* [42] and Ni *et al.* [43] emphasized the viscosity-temperature relation in their slag flow models. For instance, Ni *et al.* considered the ash flow temperature for the transition between the immobile and flowing slag layers. However, flowing slag between the flow temperature and critical viscosity temperature had a different viscosity-temperature relation than flowing slag above the critical viscosity temperature.

Yong [33] combined Seggiani's [39] slag flow model with momentum transfer of captured particles introduced by Wang *et al.* [35]. An analytical steady-state model was

produced by developing an energy balance for the reactor wall and slag layers. Furthermore, the linear temperature profile assumed by Seggiani was replaced by a cubic temperature profile. However, slag properties, including viscosity, are treated as constants in the flowing slag layer.

Validation of slag flow models is extremely limited. One reason for this is that gasifier vendors and the user community are very protective with information on process design and operation, limiting the data available publicly [44]. Another reason is that experimental data near the slag layer are very difficult to obtain due to harsh conditions in gasifiers [33,35]. An alternative may be to validate slag flow models indirectly with outer wall temperatures and heat loss to a coolant since slag buildup affects heat transfer to the reactor walls. The slag mass flow rate exiting the gasifier may also be a useful indicator. Li *et al.* [42] performed a quench of their pilot-scale gasifier at the end of a run and measured the slag layer thickness at four locations. Values ranged from 6.38-6.81 mm, while their slag flow model prediction was 6.21 mm at these locations. Obviously much more data of this type are needed for proper validation.

Many aspects of current slag flow models can be improved. All models mentioned above assume a thin slag layer (relative to the reactor radius) allowing the use of Cartesian coordinates. However, the slag layer is not always thin and there is much interest in simulating cases where the reactor plugs or is nearly plugged due to slag buildup. Hence, it may be necessary to use cylindrical coordinates, starting with the momentum continuity equation. Moreover, some models provide an analytical solution while others provide a numerical scheme. The advantage of an analytical solution may be apparent, such as faster simulation due to lesser calculations, and an exact result provided the assumptions are valid. On the other hand, a numerical solution may provide more flexibility with various parameters and limit the assumptions required. For example, Seggiani's model discussed above requires a viscosity-temperature equation which can be integrated and the overall model can change significantly if another viscosity-temperature equation is used. As for choosing between a steady-state or transient slag flow model, the choice will depend upon the specific application. Steady-state calculations are simpler, require fewer

assumptions, and may provide all the desired information. Conversely, some gasifier operating conditions may never reach steady-state between start-up and shut-down. Finally, various equations for specific phenomena will have to be added to the model to accommodate various conditions. For example, some gasifiers operate with very high gas flow rates and the shear stress on the slag surface is non-negligible. This will affect the boundary conditions when solving the slag momentum balance and a gas shear stress equation will have to be included.

## **2.6 Slag-refractory interactions**

During operation, the inner walls of entrained-flow slagging gasifiers are exposed to harsh conditions such as elevated temperature (as high as 1600°C), elevated pressure (20 bar or higher) and contact with corrosive and erosive gases, liquids and solids [45-47]. Furthermore, these parameters can vary significantly over the gasifier lifetime, particularly during start-up and shut-down. Gasifiers are typically classified as having either a refractory lining design or a water-cooled membrane design [48]. The refractory lining design is simpler and less expensive. Refractories are usually alumina-based (sintered and/or fused cast alumina-silicate, high alumina, chromia-alumina, chrome-magnesia spinels, alumina-chromia-zirconia), provide good thermal insulation and low energy losses [46]. However, their lifetime is usually only 3-30 months. Replacement cost and downtime losses are significant. In comparison, membrane walls typically consist of high-pressure tubes, in which steam is generated, covered by a castable refractory (usually SiC) [48]. Ideally, a solid slag layer forms on the wall and protects it. Good service life can be obtained (> 8 years). However, membrane walls are expensive and experience more heat loss than refractory walls. Regardless of the type of wall used, slag has a huge impact on the performance and lifetime of the wall. Basic slag-refractory interaction concepts and models are discussed in the following. Note that the “refractory” in question may be that of a refractory lining design, or a water-cooled membrane design.

Rawers *et al.* [45] lists eight mechanisms by which slag can degrade refractory: (1) refractory dissolution; (2) refractory particles debonding; (3) slag-refractory elemental replacement; (4) slag-refractory reaction producing new phases; (5) slag-refractory physical mismatch; (6) slag penetration into refractory pores; (7) slag wear-abrasion-erosion; (8) stress (and thermal cycle) loadings resulting in fragmentation and spallation. Much of the experimental data studying these mechanisms derive from cup tests during which slag is placed in cups of various refractory materials at relevant temperatures [45,46]. After a given amount of time, the slag, refractory and slag-refractory interface are analysed. Nakano *et al.* [49] and Park *et al.* [50] studied slag-refractory interactions by sessile-drop type experiments using a confocal scanning laser microscope. Quenched samples were analysed by scanning electron microscope couple with energy-dispersive X-ray spectroscopy (SEM-EDX).

Thermodynamic calculations are often relied upon to model refractory corrosion by slag. Park *et al.* [50] used the thermodynamics software FactSage [51] to determine the amount of CO and SiO gases formed via reaction between slag and a SiC refractory. Kovacik *et al.* [52] used FACT (which is now part of the FactSage software package) to study the dissolution of Cr<sub>2</sub>O<sub>3</sub>-Al<sub>2</sub>O<sub>3</sub> refractory in various slags with various gas atmospheres. The solubility limit of refractory in slag was predicted by progressively adding larger amounts of refractory to the slag and gas until the slag could no longer incorporate more chromium and chromium solids were stable at equilibrium. It was noted that the activity of each phase, as calculated by FACT, can be used to calculate the thermodynamic driving force of phase formation. Luz *et al.* [53] also used FactSage to predict slag-refractory interactions. However, their focus was that of slag penetrating refractory. 100 g of refractory were repeatedly added to the equilibrium liquid component of slag from the previous step, thus modeling liquid slag transformation as it penetrates the refractory. Predictions compared well to results from cup tests.

Still, degradation of refractory by interaction with slag is not solely a chemical process. Many physical aspects play a key role. Lin *et al.* [54] developed a transient numerical model for simulation of thermal stresses in a slag layer built up on a water-cooled

membrane wall. The effects of slag thickness and porosity were investigated. Johnson *et al.* [55] developed a refractory degradation model using the finite element method to predict the combined process of slag infiltration and spalling. The slag concentration field in the refractory is solved incrementally so that effective diffusivity (determined experimentally) can be input as a temperature-dependent quantity. The swelling strain was defined as a linear function of the fractional slag concentration. The thermal strains were based on a linear thermal expansion coefficient and the temperature change from an initial reference temperature. The refractory elastic modulus, critical cracking stress, and softening modulus (all three taken from literature) were used to predict refractory failure. Various sensitivity analyses were performed. Model results vaguely compared well with previously reported operation observations. Williford *et al.* [56] developed two models for high-chromia refractory spalling in slagging gasifiers. The first model treats spalling due to compressive forces caused by the volumetric expansion associated with the replacement of chromium by slag-borne iron. As the slag penetrates the porous refractory, an irregular diffusion front is established and slag iron reacts with refractory chromium. The difference in growth strain between the slag-affected refractory and host refractory leads to cracking and eventually spalling. The second model treats spalling due to tensile forces caused by the volumetric shrinkage associated with chromium volatilization and loss from the refractory matrix. Both models make use of slag effective diffusivity, which in turn is a function of surface tension, contact angle and viscosity. Williford *et al.* note that there are few quantitative models in the literature specifically designed for prediction of gasifier refractory spalling, and even fewer quantitative spalling data. They suggest spall thickness as an indicator of spalling mechanism, since the compressive stress model predicts a noticeably thicker spall (~4 cm) than the tensile stress model (~1.3 cm).

Refractory failure alone is one of the greatest concerns for gasifier operators. Slag-refractory interaction models are relatively new and will require many more studies. Furthermore, interaction between slag and refractory should eventually be incorporated in comprehensive gasifier models as some inputs for slag-refractory interaction models will come from other sub-models, and vice versa. For example, dissolution of refractory

components in slag can greatly impact slag flow. Slag-refractory interaction models presented above tend to either focus on corrosion (i.e., thermodynamic models) or physical degradation (i.e., stress-induced spalling models). Close coupling of these two approaches could lead to a far superior model. Experimental data for validation will be crucial, despite technical difficulties such data collection may present.

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## **Chapter 3. Artificial neural network model to predict slag viscosity over a broad range of temperatures and slag compositions**

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### **3.1 Abstract**

Threshold slag viscosity heuristics are often used for the initial assessment of coal gasification projects. Slag viscosity predictions are also required for advanced combustion and gasification models. An artificial neural network model was developed to predict slag viscosity over a broad range of temperatures and slag compositions. This model outperforms other slag viscosity models, resulting in an average absolute logarithmic error of 0.703 when applied to validation data. Genesee coal ash viscosity predictions were made to investigate the effect of adding Canadian limestone and dolomite. The results indicate that magnesium in the fluxing agent provides a greater viscosity reduction than calcium for the threshold slag tapping temperature range.

*Keywords:* Slag, Viscosity, Artificial neural network, Model

## 3.2 Introduction

Slag viscosity has major implications in metallurgical processes, coal combustion and coal gasification. When coal is combusted or gasified at a sufficiently high temperature, its ash component will form slag which can damage surfaces, reduce heat exchange and create fouling. In entrained-flow slagging gasifiers, some slag build-up on the reactor's inner surface is desired to protect refractory lining and reduce heat loss. However, a great concern in gasification design is plugging by slag, with an often quoted maximum slag viscosity of 25 Pa·s for proper operation [1,2]. To ensure slag tappareability, one option is to raise the gasifier operating temperature, which can significantly increase oxygen demand and overall cost, whereas another option is to blend fuels or add a fluxing agent. Possible fluxing agents include limestone and dolomite due to their availability, low cost and elevated content in calcium and magnesium, which generally reduce viscosity. Front-end engineering and design (FEED) work is being conducted on a 270 MW Integrated Gasification Combined Cycle (IGCC) power plant to be constructed in Alberta, Canada [3]. For this work, 18 tonnes of Genesee coal were sent to the selected gasifier vendor to test their technology with Albertan coal. Viscosity predictions for typical gasifier conditions and slag compositions can be used for initial feasibility assessments, while advanced combustion and gasification computational fluid dynamics models require local slag viscosity predictions to estimate sticking probabilities and flow throughout the reactor [4-6]. Since operating conditions and slag compositions can vary significantly not only between processes, but also within a given process, an accurate viscosity model with a wide range of applicability is desired. The objectives of this study are to develop and apply such a model with a test coal ash, here Genesee coal ash, with varying amounts of fluxing agents.

## 3.3 Slag viscosity

### 3.3.1 Parameters affecting viscosity

Like any fluid, slag rheology is a function of its molecular structure and dynamics. Slag usually has silicon dioxide as its main component and can thus be categorized as a silicate melt for which the random network model is often used to describe its rheology [2,7]. According to this model, silicate melt components can be classified into three categories; network formers, network modifiers and amphoteric. Network formers (e.g.,  $\text{Si}^{4+}$  and  $\text{Ge}^{4+}$ ) stabilize the network and therefore increase viscosity. Network modifiers (e.g.,  $\text{Na}^+$ ,  $\text{K}^+$ ,  $\text{Mg}^{2+}$ ,  $\text{Ca}^{2+}$ ,  $\text{Fe}^{2+}$ ,  $\text{Cr}^{3+}$ ,  $\text{V}^{5+}$ ,  $\text{Ba}^{2+}$  and  $\text{Sr}^{2+}$ ) disrupt the network and therefore decrease viscosity. Amphoteric (e.g.,  $\text{Al}^{3+}$ ,  $\text{Fe}^{3+}$ ,  $\text{B}^{3+}$  and  $\text{Zn}^{2+}$ ) can act either as network formers or modifiers. Although these general tendencies often hold true, the actual phenomena is more complex leading to discrepancies between this description and measured viscosities. Some discrepancies may be explained by changes in oxidation level, changes in coordination, and phase separation. For instance, ferrous iron ( $\text{Fe}^{2+}$ ) is considered to be a network modifier, while ferric iron ( $\text{Fe}^{3+}$ ) is considered to be amphoteric. The oxidation state of iron will depend upon temperature, pressure and the gaseous atmosphere above the slag [8]. Moreover, aluminum tends to be tetrahedrally coordinated and act as a network former when its charge can be balanced by alkali metal or alkaline earth metal cations [9]. If its charge is not balanced by a cation, it tends to be hexahedrally coordinated and weaken the silicate melt structure.

Slag viscosity also has a strong dependence on temperature. At elevated temperatures, slag acts as a Newtonian fluid and its viscosity usually decreases logarithmically with increasing temperature. At lower temperatures, crystallization or the separation of immiscible liquids may occur causing a dramatic increase in viscosity and non-Newtonian behaviour.

Although there is no doubt that the atmosphere surrounding a silicate melt can impact its viscosity [2,10], the relationship between the two is unclear. Under reducing conditions,

the viscosity at a given temperature is generally lower than in air. The effect seems to be greater in silicate melts with high iron content, potentially due to a higher proportion of iron in the ferrous state. The presence of water was also shown to lower or increase viscosity depending on the melt's composition.

Finally, the measurement technique can also affect the observed viscosity values. Vargas *et al.* [7] reviewed various types of viscometers and concluded that rotational viscometers are best suited for silicate melts mainly due to their range and accuracy. For this type of viscometer, only the spindle and the crucible come into contact with the melt. Material selection for these pieces is difficult since few materials can withstand the high temperatures without contaminating the sample. Vargas *et al.* recommended that platinum-rhodium be used for oxidizing or inert conditions, and molybdenum be used for reducing or inert conditions.

### **3.3.2 Previous models**

Most models developed to predict the viscosity of silicate melts are semi-empirical models based on theoretical relations describing Newtonian fluids and relying on data regression to obtain some empirical parameters. Example of these include the  $S^2$  [1], WF [1], Urbain [11], Riboud [12] and KF [13] models. Vargas *et al.* [7] tested these and more for silicate melts containing 20-75 mol% silicon dioxide, with temperatures in the range of 1550-1900 K and measured viscosities between 0.01-10,000 Pa·s. Limited success was obtained, with predictions often in error by several orders of magnitude. However, Vargas *et al.* stated that for these tests, no attempt was made to maintain within the compositional ranges for which the models were intended. Browning *et al.* [1] also developed a semi-empirical model (BBHLW model) for Newtonian fluids in which it is assumed that at a given viscosity the gradient of the viscosity-temperature curve is the same for all coal slags. A linear regression was performed to obtain an equation relating viscosity to a modified temperature. The modified temperature is the difference between the actual temperature and a shift temperature, which is in turn correlated to composition

by an empirical equation. For coal and synthetic slags with viscosities less than 1000 Pa·s, their model outperformed the  $S^2$ , KF and WF models.

Semi-empirical models relating the viscosity of non-Newtonian (heterogeneous) melts to the viscosity of Newtonian (homogeneous) melts are also available [7]. For silicate melts, non-Newtonian behaviour can be caused by the appearance of crystals in the melt, or the separation of immiscible liquids. Most non-Newtonian models require knowledge of the crystal fraction and sometimes crystal shape or size. F\*A\*C\*T (now part of the FactSage™ software package) can be used to determine phase fractions and compositions for adjustment of the viscosity prediction [14]. Dyk *et al.* [15] employed FactSage to estimate the composition of the liquid portion of heterogeneous slags and used this composition in the KF model. It was stated that this method is highly likely to be more accurate, but without providing quantitative details. Kondratiev and Jak [16] optimized the Urbain model with experimental data of homogeneous melts as determined by F\*A\*C\*T. To calculate the viscosity of a heterogeneous slag, the proportion of solids and the composition of the remaining liquid phase were first estimated using F\*A\*C\*T, then the liquid phase viscosity was calculated from the optimized Urbain model, and finally this viscosity was adjusted depending on the volume fraction of solid particles, according to the Roscoe equation (given in [16]). Again without giving quantitative details, it was stated that good agreement was achieved between the model and over 4000 experimental points. In sum, the integration of FactSage calculations when predicting slag viscosity shows promise, but published quantitative comparison of predictions and experimental values are lacking. Also, although the FactSage database is constantly expanding, limitations may still exist with certain compositions where higher order interactions are encountered.

Artificial neural networks (ANNs) are gaining popularity in various engineering domains. They attempt to reproduce the functionality of the brain which consists of many interconnected neurons functioning in parallel. Each neuron will receive signals of varying intensities, process them, and accordingly send out a signal with a given intensity to be processed by other neurons. ANNs present significant advantages over conventional

data regression models [17] as there is no need to provide a model function, the learning capability of ANNs allows the discovery of more complex and subtle interactions between variables, and ANNs are intrinsically more robust when it comes to handling noisy or inaccurate data. In designing an ANN, it is important to consider the structure of the network, as well as the learning algorithm used to train it. To train an ANN, sets of inputs and corresponding outputs are provided, and the weights are initialized to random values. Inputs are sequentially sent through the ANN and the error on the calculated output is determined from the provided output sets, according to a specified performance function. The weights of the signals are adjusted to minimize the error. The process is repeated until a certain number of iterations are completed or a given error tolerance is attained. Other stop criteria may be specified. The method by which weights are adjusted depends on the learning algorithm. ANNs should be validated with data which was not used for training in order to detect over-fitting. They are particularly prone to over-fitting due to the large number of parameters used. Reviews covering various aspects of neural network modeling are available in [17,18].

A literature search revealed two studies describing silicate melt viscosity modeling via ANNs. The earliest attempt was based on 350 data points of magnesium-rich slags and the developed ANN was shown to outperform a linear regression model, prepared by the authors, which was based on the same data [19]. Hanao *et al.* [20] produced neural networks for mould fluxes (a type of silicate melt) using data collected during a round robin project involving 21 research groups. They compared their neural network predictions with those of the Iida model [21] and stated that the neural networks performed much better without providing quantitative details. However, comparisons were made over a very small range (0-6 Pa·s) of viscosities. Although these two attempts to model silicate melt viscosity by ANNs are promising, their range of applicability may be highly constrained in terms of composition (magnesium-rich slags in the first study) and viscosity (below 6 Pa·s in the second study). Hence, in this study it was desired to develop a general slag viscosity model with a wide range of applicability.

## 3.4 Artificial neural network modeling of slag viscosity

### 3.4.1 Dataset and input variable selection

There are numerous silicate melt viscosity measurements available in the literature. Unfortunately, the techniques used to obtain them vary greatly and the sources of potential errors are numerous. In addition, many different components are found in silicate melts and creating a model that accounts for all of them may be unrealistic. To control some of the variability caused by experimental technique, all data used in this study was obtained using a rotational viscometer, with the exception of viscosity data from dataset U (see Table 3.1) in the range of  $10^7$  to  $10^{12}$  Pa·s, which was obtained by a falling film method. Although it would be desirable to also limit the selected datasets according to spindle and crucible materials used, a great deal of data is required to create an ANN with a wide range of applicability, and such material stringency would eliminate too much data from the study. Nevertheless, the authors believe that the effect caused by spindle/crucible contamination is negligible compared to the effect of the primary input sources of the model. An attempt was made to account for atmosphere used during measurements. None of the conventional regression models mentioned previously directly account for the slagging atmosphere (or its pressure). Some may indirectly account for atmosphere by differentiating ferric and ferrous iron, but this introduces errors associated with determining the quantity of each ion present in the melt. With neural networks, the atmosphere can be accounted for by one or more additional inputs.

A few trials were performed using an additional input to those listed below, with air being attributed a value of 1, and nitrogen and argon a value of -1. Data that was not obtained with one of these atmospheres, or for which the atmosphere is unknown, was not used. Although one of the ANNs developed performed very well, when varying the initial weights prior to training, the results obtained were inconsistent. This suggests that there may be over-fitting. ANNs developed while limiting the data to one atmosphere had poor performance, likely due to lack of data. For these reasons, atmosphere was not controlled for the final ANN. Table 3.1 lists the datasets used in this study, the number of

data points in each, as well as the spindle/crucible material and atmosphere used for measurements.

**Table 3.1: Datasets used for ANN models in this study**

Letter code	Reference	Material	Atmosphere	Data points
A	Persson (2007) [22]	Fe/Mo	Ar	111
B	Shankar <i>et al.</i> (2007) [23]	Mo	Ar	54
C	Hurst <i>et al.</i> (2000) [24]	Mo	N <sub>2</sub>	51
D	Hurst <i>et al.</i> (1999a) [25]	Mo	N <sub>2</sub>	105
E	Hurst <i>et al.</i> (1999b) [26]	Mo	N <sub>2</sub>	170
F	Hurst <i>et al.</i> (1996) [7]	Mo	N <sub>2</sub>	18
G	Stein and Spera (1993a) [7]	Al <sub>2</sub> O <sub>3</sub>	air	103
H	Stein and Sperra (1993b) [7]	Al <sub>2</sub> O <sub>3</sub>	air	7
I	Nowok and Benson (1991) [7]	unknown	CO/CO <sub>2</sub>	12
J	Dingwell (1989) [7]	Pt/Rh	air	23
K	Dingwell and Virgo 1988 [7]	Pt/Rh	air	80
L	Jones and Lindsey (1987) [7]	unknown	unknown	63
M	Scarfe and Cronin 1986 [7]	Pt/Rh	unknown	42
N	Vorres <i>et al.</i> (1986) [7]	Al <sub>2</sub> O <sub>3</sub> /Mo	H <sub>2</sub> /CO/N <sub>2</sub> , air	16
O	Mysen <i>et al.</i> (1985) [7]	Pt/Rh	air	16
P	Quon <i>et al.</i> (1985) [7]	Mo	unknown	106
Q	Hochella and Brown (1984) [7]	unknown	unknown	6
R	Seki and Oeter (1984) [7]	Pt/Rh	air, CO <sub>2</sub>	49
S	Scarfe 1983 [7]	Pt/Rh	air	75
T	Schobert <i>et al.</i> (1982) [7]	C/Al <sub>2</sub> O <sub>3</sub>	H <sub>2</sub> /N <sub>2</sub> , N <sub>2</sub> , air	37
U	Urbain <i>et al.</i> (1982) [7]	Mo/W	vacuum, Ar	482
V	Piwinskii and Weed (1980) [7]	unknown	air	37
X	Segers <i>et al.</i> (1979) [7]	Pt/Rh	Ar	18
Y	Bodnar <i>et al.</i> (1978) [7]	Fe/Pt	N <sub>2</sub>	163

The inputs selected for the ANN are elemental mole fractions and temperature, in kelvins, normalized as follows:

$$temperature\ input = \frac{temperature - 1082}{2755 - 1082}$$

Oxygen and components which never exceed a mole fraction of 0.02 in the selected datasets are not considered here. Mole fractions were normalized to have a sum of 1. The ranges of each input when combining all datasets used are given in Table 3.2.

**Table 3.2: Input and output ranges of ANN models in this study**

<b>Input/output</b>	<b>Minimum</b>	<b>Maximum</b>
Temperature (K)	1082	2755
Si mole fraction	0.13	1.00
Al mole fraction	0.00	0.82
Fe mole fraction	0.00	0.81
Ca mole fraction	0.00	0.60
Mg mole fraction	0.00	0.65
Na mole fraction	0.00	0.50
K mole fraction	0.00	0.51
Mn mole fraction	0.00	0.69
Log(viscosity (Pa·s))	-3.81	11.16

### **3.4.2 Performance function**

The proper selection of the function which is minimized when adjusting the parameters of a model, the performance function, is important. One performance function which is commonly used is the average absolute error (*AAE*):

$$AAE = \frac{1}{N} \sum_{i=1}^N |p_i - m_i|$$

where  $p_i$  is the  $i^{th}$  predicted value and  $m_i$  is the  $i^{th}$  measured value for  $N$  data points. The trouble with using  $AAE$  is that errors for small values, which may be relatively large, are summed with errors on large values. Usually there is a greater tolerance for error (difference between the prediction and measurement) when values get larger. The same problem is seen with the average squared error ( $ASE$ ). To account for this greater error tolerance on larger values, the average absolute relative error ( $AARE$ ) is often used:

$$AARE = \frac{1}{N} \sum_{i=1}^N \left| \frac{p_i - m_i}{m_i} \right|$$

This will result in a value which is plus or minus a given percentage of the measured value. Unfortunately, this function does not differentiate a positive relative error from a negative relative error. For instance, the model user will likely prefer an error of +60% (which yields a prediction which is off by a factor of 1.6) than an error of -50% (which yields a prediction which is off by a factor of 2). The  $AARE$  performance function will favour the inverse. Hence a performance function which gives an average factor difference would be best. This is achieved by using the average absolute logarithmic error ( $AALe$ ):

$$AALe = \frac{1}{N} \sum_{i=1}^N \left| \log_{10}(p_i) - \log_{10}(m_i) \right|$$

where, on average, the measured value is equal to the predicted value multiplied or divided by 10 to the power of  $AALe$ . The average squared logarithmic error ( $ASLE$ ) will yield similar results.

### **3.4.3 Architecture and training**

ANNs were created using MATLAB<sup>®</sup> version 7.6 [27]. All ANNs tested were feed-forward with back-propagation. Datasets C, D, L, P and X were used for validation

whereas all other datasets were used for training. Entire datasets were used for validation to better detect possible over-fitting. If random data points or compositions had been used for validation, the training set would likely have contained very similar data, rendering it difficult to determine how well the ANN would perform on a new set of data. This would create a bias when comparing the ANN to conventional regression models. Furthermore, to reduce the likelihood of over-fitting the data, the *trainbr* training function was used. This function applies the Levenberg-Marquardt algorithm with Bayesian regularization [27]. The *divideFcn* function was given a blank parameter to deactivate early stopping (a default generalization method which is not necessary when Bayesian regularization is used). The output values for training were logarithm (base 10) values of the viscosity in Pa·s. All other settings were left as the defaults of the *newff* and *train* functions which were used for ANN creation and training, respectively. By default the inputs are normalized to values between -1 and 1, the hidden nodes are tan-sigmoid transfer functions, and the output node is a linear transfer function. Tests were conducted varying the number of hidden layers and the number of nodes per hidden layer. A single hidden layer with 3 nodes provided the best results.

### **3.4.4 Comparison of models**

The Urbain, Riboud, KF and BBHLW models are compared to three ANNs in Table 3.3. The only difference between the three ANNs is the initial weights used prior to training. The *AALE* of each is shown for all, training and validation data. Dataset U was not used in this analysis because the BBHLW model was unable to return a value for some of the compositions. This was due to a division by 0 in one of the equations. However, all three ANNs outperform the Urbain, Riboud and KF models with dataset U. Of the four conventional semi-empirical models, the BBHLW model is the best with the training data, achieving an *AALE* of 0.951, which corresponds to a multiplication/division error factor of 8.93. All ANNs outperform the BBHLW model with the training data. ANN1 is the best, achieving an *AALE* of 0.663, which corresponds to an error factor of 4.60. Of the four conventional models, the Riboud model is the best with the validation data achieving an *AALE* of 0.918, which corresponds to an error factor of 8.28. All ANNs

outperform the Riboud model with the validation data. ANN1 is the best, achieving an *AALE* of 0.703, which corresponds to an error factor of 5.05. Overall, ANN1 has the best performance and its final weight values are given in Table 3.4. For the ANNs, the small difference in *AALE* between the training and validation sets, as well as between each other, are indications of good generalization.

**Table 3.3: Average absolute logarithmic errors of compared models for each dataset group**

<b>Model</b>	<b>Dataset group</b>		
	<b>All</b>	<b>Training</b>	<b>Validation</b>
Urbain	1.164	1.191	1.083
Riboud	1.093	1.152	0.918
KF	1.134	1.182	0.992
BBHLW	0.954	0.951	0.964
ANN1	0.673	0.663	0.703
ANN2	0.734	0.699	0.837
ANN3	0.714	0.684	0.803

**Table 3.4: ANN1 weight values**

<b>Weights to 1st hidden layer</b>			
<b>From/To</b>	<b>Node 1</b>	<b>Node 2</b>	<b>Node 3</b>
1 (Temperature)	-0.7939	1.1056	-0.1516
2 (Si)	1.1249	0.1558	-0.4868
3 (Al)	-0.5146	-1.7803	0.9970
4 (Fe)	0.0577	0.6035	-0.2673
5 (Ca)	0.7364	0.3918	-0.5305
6 (Mg)	-1.6109	-1.1210	1.3102
7 (Na)	0.6952	0.1849	-0.4222
8 (K)	0.2251	0.9112	-0.4758
9 (Mn)	0.8590	1.4353	-1.0365
Bias	-0.2809	-0.0547	0.1204

<b>Weights to output</b>	
<b>From/To</b>	<b>Output</b>
Node 1	1.3008
Node 2	1.0609
Node 3	2.2571
Bias	-0.2587

### **3.5 Genesee coal ash viscosity predictions with varying amounts of fluxing agents**

A Genesee coal ash sample composition is given in Table 3.5. The composition of two Canadian limestone samples (FA1 and FA2) and a Canadian dolomite sample (FA3) are given in Table 3.6. Slag viscosity predictions were made via ANN1 assuming that the slag composition is uniform and that the magnesium carbonate and calcium carbonate are respectively converted to magnesium oxide and calcium oxide, with carbon dioxide going

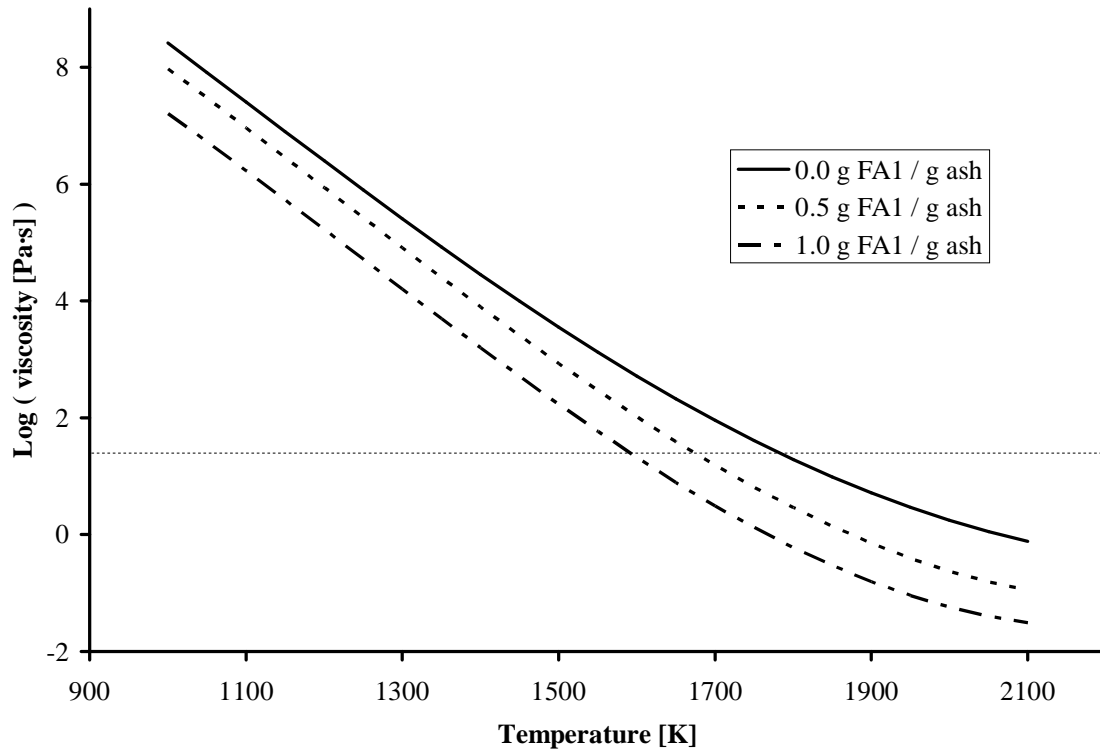
to the gas phase. Figures 3.1-3.3 show the effect of adding FA1-FA3, respectively. The results suggest that if no fluxing agent is added, gasifier operation would have to be kept above 1750 K to have the slag viscosity below the recommended maximum of 25 Pa·s. Addition of 0.5 g FA1 per 1 g ash reduces the required temperature to ~1675 K, while addition of 1 g FA1 per 1 g ash further decreases the required temperature to ~1600 K. FA2 additions yield similar results, although the required temperatures are slightly higher, likely due to greater silicon dioxide content. Conversely, the effect of FA3 addition is much more pronounced. Addition of 0.5 g FA3 per 1 g ash reduces the required temperature to ~1500 K while addition of 1 g FA1 per 1 g ash further decreases the required temperature to ~1350 K. FA1 and FA3 are very similar in composition except for their calcium carbonate and magnesium carbonate fractions. Although they have roughly the same amount of combined calcium carbonate and magnesium carbonate, FA3 has much more magnesium carbonate. Hence, with this ash, magnesium seems to be a better fluxing agent than calcium. Browning *et al.* [1] also predict a greater viscosity reducing effect with magnesium than with calcium in their model. Furthermore, the addition of FA3 seems to reduce the slag's sensitivity to temperature at higher temperatures, as can be seen by the reduction in slope with increasing temperature in Figure 3.3. Consequently, if the viscosity threshold is set to a lower limit, or gasifier operation is conducted at a higher temperature, addition of a magnesium-rich fluxing agent may not be favourable. Finally, it must be stressed that these predictions should only be taken as a preliminary basis for gasifier design while general trends are investigated. ANN1 has an *AALE* of 0.703 for data it was not trained with, meaning that the actual viscosity could easily be 5 times greater or smaller. Viscosity measurements closer to the desired composition, temperature and atmosphere would provide indispensable information.

**Table 3.5: Genesee coal ash composition (adapted from [28])**

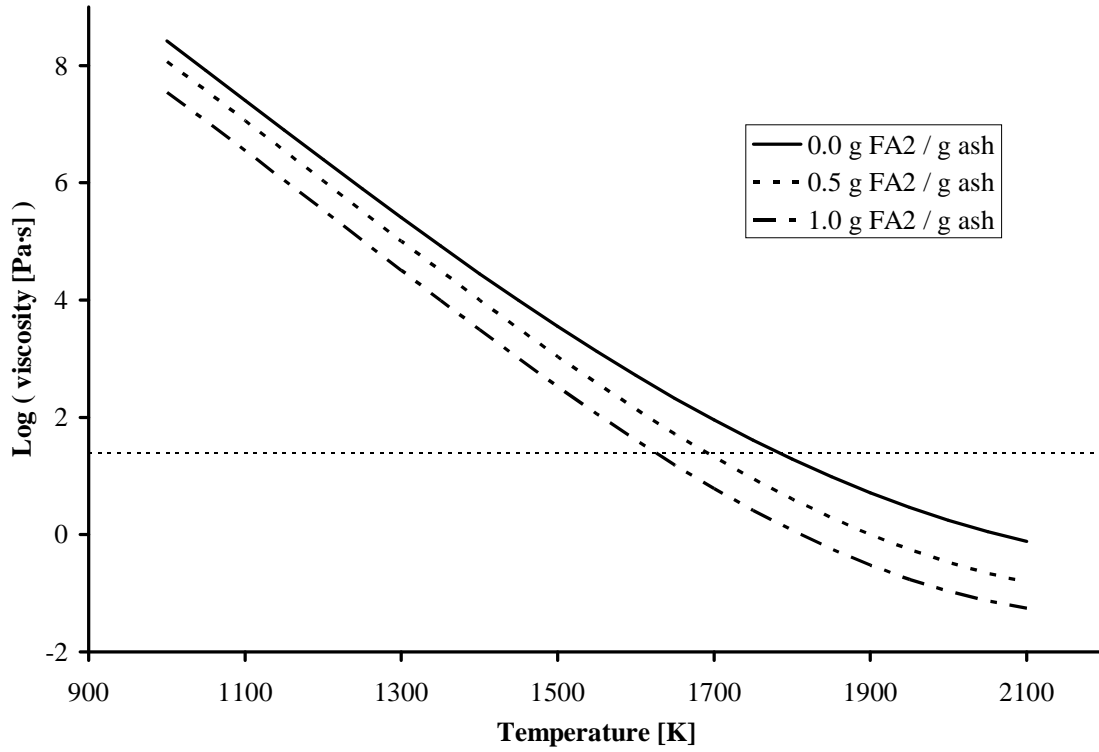
<b>Component</b>	<b>Weight%</b>
SiO <sub>2</sub>	59.2
Al <sub>2</sub> O <sub>3</sub>	19.8
Fe <sub>2</sub> O <sub>3</sub>	3.81
TiO <sub>2</sub>	0.6
P <sub>2</sub> O <sub>5</sub>	0.04
CaO	5.0
MgO	1.35
Na <sub>2</sub> O	2.18
K <sub>2</sub> O	1.41
BaO	0.27
SrO	0.07
MnO	0.04
V <sub>2</sub> O <sub>5</sub>	0.03
SO <sub>3</sub>	2.21

**Table 3.6: Compositions of fluxing agents**

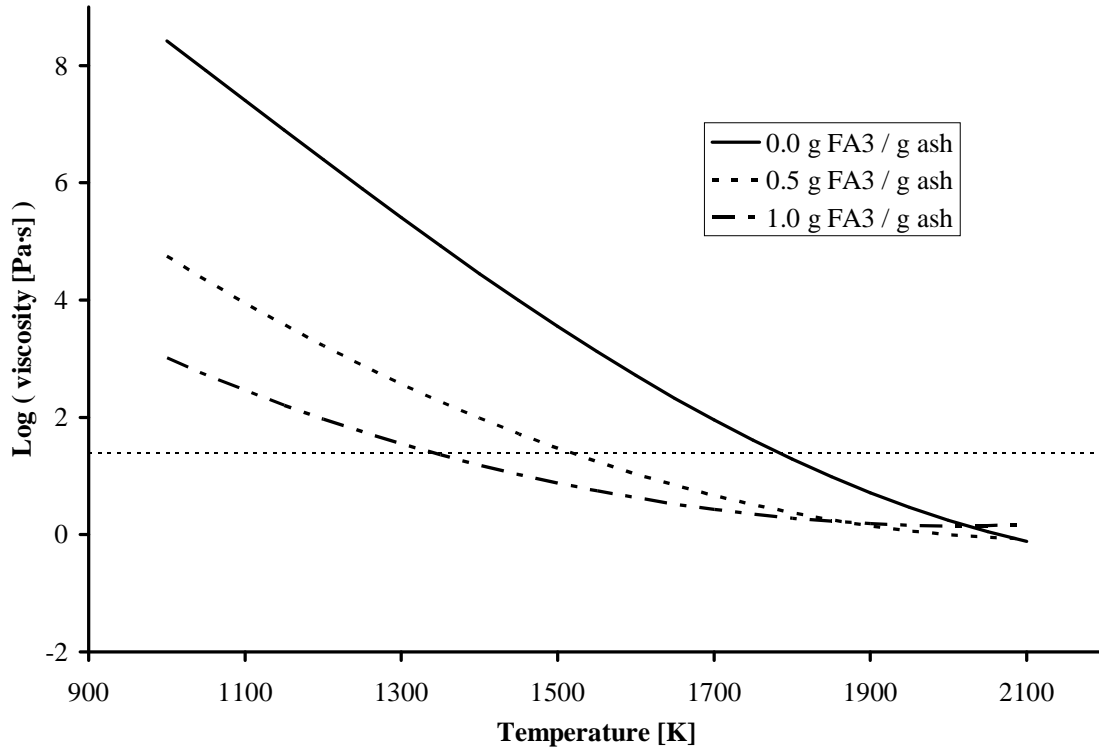
<b>Component</b>	<b>Weight%</b>		
	<b>FA1</b>	<b>FA2</b>	<b>FA3</b>
CaCO <sub>3</sub>	91.3	75.6	50.6
MgCO <sub>3</sub>	2.49	2.50	40.4
Fe <sub>2</sub> O <sub>3</sub>	1.00	1.59	0.94
Al <sub>2</sub> O <sub>3</sub>	1.34	3.93	0.98
SiO <sub>2</sub>	5.98	14.4	6.38
Na <sub>2</sub> O	0.03	0.04	0.16
K <sub>2</sub> O	0.23	0.93	0.02



**Figure 3.1: Viscosity predictions for Genesee coal ash with varying amounts of FA1. The horizontal hashed line corresponds to a viscosity of 25 Pa·s.**



**Figure 3.2: Viscosity predictions for Genesee coal ash with varying amounts of FA2. The horizontal hashed line corresponds to a viscosity of 25 Pa·s.**



**Figure 3.3: Viscosity predictions for Genesee coal ash with varying amounts of FA3. The horizontal hashed line corresponds to a viscosity of 25 Pa·s.**

### 3.6 Conclusion

An ANN was created to predict slag viscosity over a wide range of temperatures and slag compositions. Various measures were taken to avoid over-fitting the training data. The ANN was shown to outperform other models with an *AALE* of 0.703 when applied to validation data. Slag viscosity predictions for Genesee coal ash were evaluated to determine the effect of various fluxing agents. Of the fluxing agents studied, the one with high magnesium content has the greatest effect when it comes to minimizing the required temperature for slag removal.

### 3.7 Acknowledgements

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## Chapter 4. Slag Viscosity Modeling Toolbox

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## 4.1 Abstract

Slag viscosity models are applied in many industries. However, the models are only applicable to a limited range of slag compositions and conditions, and their performance is not easily assessed. The present study describes tools that have been developed to assist slag viscosity model users in the selection of the best model for given slag compositions and conditions, and to help users determine how well the model will perform. The tools, which are in the form of several publicly available files and programs, include a slag viscosity prediction calculator with 24 slag viscosity models, and a database of 4124 slag viscosity measurements. The database includes over 750 compositions from 53 published studies. New slag viscosity models, integrated into the tools, include an artificial neural network for fully molten slags, and a viscosity prediction modifier for slags containing solid particles. Glass forming, entrained flow gasification and blast furnace case studies are provided to demonstrate how the slag viscosity modeling tools can be applied and to highlight certain features that should be considered when using slag viscosity models and experimental data.

Keywords: Slag, Viscosity, Database, Model.

## 4.2 Acronyms

AAE	average absolute error
AALE	average absolute logarithmic error
AARE	average absolute relative error
ANN	artificial neural network
CFD	computational fluid dynamics
RE	Roscoe-Einstein
SVD	SlagViscosityDatabase tool
SVP	SlagViscosityPredictor tool

## 4.3 Introduction

The viscosities of slags have been widely investigated over the past five decades. Slag viscosity has major implications in glass production, geophysical phenomena, metallurgical processes, combustion and gasification. Samples from different origins, including artificial laboratory prepared glasses [1] and ashes [2], coal fly ashes [3], magmatic systems [4, 5], gasifier slags [6], and metallurgical slags [7], have been analyzed and semi-empirical viscosity models have been developed. There is great interest in models since slag viscosity measurements are costly, lengthy to perform and require much expertise to obtain accurate data. A summary of some viscosity models has been provided by Vargas et al. [8]. With these models, viscosity is a function of the temperature and composition of the slag. However, it should be stressed that these models are optimized for limited sets of slag viscosity measurements. The models often perform very poorly with other sets of slag viscosity measurements. This discrepancy can be partially attributed to the wide range of experimental conditions used to perform slag viscosity measurements. The slag compositions, temperature range, gas atmosphere, viscosity range, heating/cooling rate, presence of multiple phases in the slag, and viscometer design vary widely from one study to the next. Furthermore, for much of the data available in the literature the slags are likely to have non-Newtonian

behavior. In these cases, the data obtained is not very useful unless the shear rate used for measurement is also known.

Slag viscosity model users may have difficulty assessing which model they should use for their particular system, and to what extent the model should be trusted. The present study describes tools that have been developed to assist in these aspects. The tools described here are being made publicly available. They consist of several files and programs available through the publisher's web site. They should be particularly useful for academic and industrial groups which do not have access to extensive private and protected slag viscosity databases and models that have been developed in-house by some organizations after many years of effort. Furthermore, it is hoped that the demonstration of how the slag viscosity data is analyzed and processed will provide guidelines on how slag viscosity data should be reported and slag viscosity models compared.

Section 4.4 describes how the slag viscosity modeling tools were developed. These tools include a slag viscosity predictions calculator and a slag viscosity database. New slag viscosity models are also described including an artificial neural network for fully molten slags, and the modification of viscosity models based on equilibrium phase predictions. Section 4.5 demonstrates how the slag viscosity modeling tools may be applied, by providing case studies from various industries which require knowledge of slag flow.

## **4.4 Calculation**

### ***4.4.1 Slag viscosity predictor tool***

The SlagViscosityPredictor tool (SVP) is used to calculate slag viscosity predictions with twenty four different models. The SVP is contained within the SlagViscosityPredictor.xls file, a MS Excel (Microsoft, Redmond, WA) file with embedded VBA macros. This file is available from the publisher's website as a supplementary file associated with this paper. The SVP tool includes reference codes and reference information for all the models that have been incorporated. The included models are the S2, Watt-Fereday, Bomkamp, Shaw, Lakatos, Urbain, Riboud, Streeter, Kalmanovitch-Frank, BBHLW, Duchesne and ANNliq models [8–10]. In addition, modified versions of all these models applying the Roscoe-Einstein model

[11] (RE modification) are also included. Details for the ANNliq and RE modification are given in Sections 4.4.3 and 4.4.4, respectively. The SVP tool provides an interface to input up to 10,000 slag temperatures and compositions. Phase information must be entered if predictions with the RE modification are desired. The crucial values for the RE modification are the masses of the (liquid) slag solution species, and the solids volume fraction.

FactSage software predicts equilibrium solid-liquid-gas phases and compositions based on Gibbs free energy minimization [12]. Gibbs free energy is calculated from optimized models with parameters based on empirical data with various compositions, temperatures and pressures. This information is contained within compound and solution databases which must be carefully selected prior to equilibrium calculations. FactSage version 6.2 was used for all phase calculations in this study. If phase predictions are requested, the SVP creates an EquiInput-FactSage.dat file which specifies the databases to use, phases to consider and pressure. The FACT53 and FToxid databases are specified, with preference given to the FToxid database in case of duplicate phases. The considered phases are listed in the SVP. A pressure of 1 atm is specified. The mass and volume fractions for liquid slag and solids are calculated by the SVP after FactSage predictions are generated. The gas phase is not considered in the mass and volume fraction calculations. The volumes of solid species are calculated using the densities obtained from FactSage. The density of the liquid slag solution is calculated from the partial molar volumes of its components. Partial molar volumes for all components considered were taken from [13], with the exception of the partial molar volume of  $K_2O$  which was taken from [14].

#### **4.4.2 Slag viscosity database tool**

The SlagViscosityDatabase tool (SVD) is used to handle slag viscosity data and evaluate slag viscosity models. The SVD is contained within the SlagViscosityDatabase.xls file, a MS Excel (Microsoft, Redmond, WA) file with embedded VBA macros. This file is available from the publisher's website as a supplementary file associated with this paper. The SVD contains 4124 measured slag viscosities from 53 references. More than 750 compositions are represented. For each slag viscosity, an identification number, temperature, molar slag composition, mass slag composition, measurement technique, gas atmosphere composition and some reference information are provided. For all slag viscosity measurements, the shear rate is marked as unknown. However, the authors wish to encourage the inclusion of this

information in future publications since it is required to model the behavior of slags with non-Newtonian behavior. The SVD lists FactSage phase predictions for all slags, as well as calculated mass and volume fractions of liquid slag and solids. These values were generated by the SVP. The SVD also lists slag viscosity predictions for all slags in the database, using 24 slag viscosity models. These were also generated by the SVP. Reference information for the measured slag viscosity data and the slag viscosity models is contained in the SVD.

The SVD provides means to identify measured viscosity data based on specified selection criteria including temperature range, composition ranges, viscosity range, measurement technique, reference information, and mass and volume fraction ranges of liquid slag and solids. When generating the dataset the SVD calculates the average absolute error (AAE), average absolute relative error (AARE), and the average absolute logarithmic error (AALE) for each slag viscosity model with the predicted and measured viscosities of the selected data. The formulas used to calculate the AAE, AARE and AALE are detailed by Duchesne et al. [10]. It is recommended that the AALE be used as an indicator of model performance. For a given model, on average, the measured value is equal to the predicted value multiplied or divided by 10 to the power of the AALE. Hence, a smaller AALE indicates better predictive performance.

#### **4.4.3 Artificial neural network model for fully molten slags**

Artificial neural networks (ANNs) are applied for modeling in various engineering domains. ANNs present significant advantages over conventional data regression models as there is no need to provide a model function, the learning capability of ANNs allows the inclusion of more complex and subtle interactions between variables, and ANNs are intrinsically more robust when it comes to handling noisy or inaccurate data. Literature reviews covering various aspects of neural network modeling are available [15, 16]. ANNs developed to model slag viscosity have been reported in the literature [10, 17, 18]. In the current study, it was desired to produce an ANN for a broad range of temperatures and compositions, but exclusively for fully molten slags. For crystal-containing slags, this model can be used in conjunction with the Roscoe-Einstein model (see Section 4.4.4).

Data for creation of the ANN for fully molten slags, dubbed ANNliq, was selected from the SVD with limitations identified hereafter. Data with a  $\text{Li}_2\text{O}$ ,  $\text{TiO}_2$ ,  $\text{B}_2\text{O}_3$ ,  $\text{P}_2\text{O}_5$ ,  $\text{NiO}$ ,  $\text{ZrO}_2$ ,

CaF<sub>2</sub>, Cr<sub>2</sub>O<sub>3</sub> or V<sub>2</sub>O<sub>5</sub> molar fraction greater than 0.01 are excluded. These components are excluded from the ANNliq model as there is limited data in the database with significant molar fractions of these components. The Yakushev 1977 dataset is excluded since, based on other viscosity measurements and model predictions, the accuracy of the viscosity values are questionable. To limit the ANNliq model to fully molten slags, data without phase information or with a liquid mass fraction lower than 1 are excluded. The inputs selected for ANNliq are temperature in kelvins and elemental mole fractions excluding oxygen. Mole fractions were normalized to have a sum of 1.

Various ANNs were created using MATLAB version R2008a [19] before obtaining the final ANNliq model. All ANNs tested were feed-forward with back-propagation. Approximately 20% of the selected data were used for validation, and the rest for training. Data that meet the requirements set above and that are from the Blomquist 1978, Bockris 1954, Bodnar 1978, Cukierman 1973, Dingwell 1988, Dingwell 1989, Duchesne 2011, Goto 1997, Hurst 1996 and Hurst 1999a references were used for validation. All other data that meet the requirements set above were employed for training. Entire datasets from given references were used for validation to better detect possible over-fitting. If random measurements had been selected for validation, the training set would likely have contained very similar data, rendering it difficult to determine how well the ANN would perform on a new set of data. This would create a bias when comparing the ANN to conventional regression models. Furthermore, to reduce the likelihood of over-fitting the data, the MATLAB *trainbr* training function was used. This function applies the Levenberg-Marquardt algorithm with Bayesian regularization. The MATLAB *divideFcn* function was given a blank parameter to deactivate early stopping (a default generalization method that is unnecessary when Bayesian regularization is used). The output values for training were logarithm (base 10) values of the viscosity in Pa·s. All other settings were left as the defaults of the MATLAB *newff* and *train* functions which were used for ANN creation and training, respectively. By default, the inputs are normalized to values between -1 and 1, the hidden nodes are tan-sigmoid transfer functions, and the output node is a linear transfer function. Tests were conducted varying the number of hidden layers and the number of nodes per hidden layer. The AALE performance function was used to evaluate and compare the models. A single hidden layer with 3 nodes provided the best results. The final weight values for ANNliq are given in Table 4.1.

The performance of the Watt-Fereday, Bomkamp, Riboud, Duchesne and ANNliq models are compared in Table 4.2. The AALE of each is shown for all selected data, training data and validation data. Other models which are available in the SVP were not included as they are unable to return a value for some of the compositions. This was due to a division by zero in one of their equations. The ANNliq model is the best with the training data, achieving an AALE of 0.25. The Watt-Fereday is the best with the validation data, achieving an AALE of 0.36. However, the ANNliq model has a similar performance, achieving an AALE of 0.38.

**Table 4.1: Weight values for ANNliq**

<b>Weights to 1st hidden layer</b>			
<b>From/To</b>	<b>Node 1</b>	<b>Node 2</b>	<b>Node 3</b>
1 (Temperature)	-0.8122	-0.6080	0.3720
2 (Si)	0.4779	1.3219	-6.4457
3 (Al)	0.2338	0.3832	-4.5697
4 (Fe)	-0.9978	-3.4329	-4.0402
5 (Ca)	0.1647	0.9149	-2.2416
6 (Mg)	0.1559	0.8349	-2.3479
7 (Na)	1.3006	2.0230	13.8132
8 (K)	0.5360	1.1477	-0.4666
9 (Mn)	-0.1862	0.3596	-1.9860
Bias	-0.2444	-0.4375	3.1203

<b>Weights to output</b>	
<b>From/To</b>	<b>Output</b>
Node 1	2.7936
Node 2	-2.3112
Node 3	-0.5613
Bias	0.3870

**Table 4.2: Average absolute logarithmic error (AALE) of slag viscosity models applied to data used for the development of the ANNliq model**

	All	Training	Validation
<b>Watt-Fereday</b>	0.63	0.69	0.36
<b>Bomkamp</b>	1.81	1.91	1.41
<b>Riboud</b>	0.48	0.49	0.44
<b>Duchesne</b>	0.62	0.63	0.58
<b>ANNliq</b>	0.27	0.25	0.38
<b>Number of datapoints</b>	1596	1284	312

#### **4.4.4 Application of the Roscoe-Einstein model for partially crystallized slags**

Most slag models are conceived for fully molten slags since their theoretical basis is not applicable to liquid-solid mixtures. The presence of solid phases in the slag will affect viscosity in two ways. First, the components which form the solids are no longer part of the liquid phase. This changes the liquid phase composition and hence its viscosity. Second, the interaction of solid particles with each other and the liquid phase will affect the bulk slag viscosity. This effect will vary with the size, shape, orientation and mass fraction of the particles. Roscoe [11] developed the following theoretical relation between the liquid phase viscosity ( $\eta_l$ ) and the solid-liquid mixture viscosity ( $\eta$ ) for a dilute solution of spheres:

$$\eta = \eta_l (1 + pf)^{-2.5}$$

where  $f$  and  $p$  are the volume fraction of solids and the solids interaction parameter, respectively. This formula was based on previous work by Einstein [20]. For a dilute concentration of spheres with diverse sizes, the solids interaction parameter takes a value of 1. With higher concentrations of spheres with similar sizes, spheres will immobilize liquid between them, increasing the apparent solid volume fraction in the mixture. In this case, the solids interaction parameter takes a value of 1.35.

In the current study, the Roscoe-Einstein model is applied in conjunction with several slag viscosity models and a solids interaction parameter value of 1. The liquid slag composition and solids volume fraction are taken from the phase information available in the SVD. The original slag viscosity models are applied to the liquid slag. The resulting viscosity prediction

is then modified according to the Roscoe-Einstein model. A similar approach was taken by Kondratiev and Jak [21]. These authors believe the Roscoe-Einstein model is applicable up to 30 vol.% solids. The performance of slag viscosity models with and without the Roscoe-Einstein modification are compared in Table 4.3. For this comparison, all data in the database for which phase information is available, and the liquid volume fraction is within 0.70-0.99, were selected. In total, 637 slag viscosity measurements were selected. Some models which are available in the SVP were not included in the comparison as they are unable to return a value for some of the compositions. This was due to a division by zero in one of their equations. Only three of the nine compared models have improved performance when the Roscoe-Einstein model is applied. The best accuracy was obtained using the Duchesne model without the Roscoe-Einstein modification. An AALE of 0.78 is obtained. The second best accuracy was obtained using the Kalmanovitch-Frank model with the Roscoe-Einstein modification. An AALE of 0.81 is obtained.

**Table 4.3: Average absolute logarithmic error (AALE) of slag viscosity models with and without the Roscoe-Einstein (RE) modification**

<b>Model</b>	<b>Without RE</b>	<b>With RE</b>
S2	0.81	1.06
Watt-Fereday	1.49	1.93
Bomkamp	2.33	3.14
Shaw	1.00	1.01
Urbain	0.90	0.92
Riboud	0.99	0.88
Kalmanovitch-Frank	0.84	0.81
Duchesne	0.78	0.85
ANNliq	1.31	1.00

## 4.5 Results and Discussion

In the following, multiple case studies are used to demonstrate how the SVD and SVP can be applied for specific slag compositions and conditions. The general strategy is to first determine which slag compositions and conditions are of most interest. The SVD is then used to identify viscosity data which match or are similar to these slag compositions and conditions. The SVD will also provide the number of viscosity measurements which match the selection criteria, and the performance of 24 slag viscosity models with these

measurements. Provided there is sufficient data which match the selection criteria, the model with the best performance is noted. Then, the specific slag compositions and conditions are entered in the SVP to generate slag viscosity predictions with the best model. Of course if the exact compositions and conditions of interest are present in the SVD, the measured viscosities can be directly used instead of relying on models. In addition to providing specific examples for application of the SVD and SVP, the case studies also demonstrate how confidence in viscosity data and model performance come into play and the delicate balance between data selection stringency and data quantity minimums for meaningful analyses. The data selection criteria for each case study are given in Table 4.4. The number of viscosity measurements identified and the performance of the slag viscosity models for each case study are given in Table 4.5.

**Table 4.4: Selection criteria for each case study (part 1 of 2)**

Case Study	Temperature (K)		Mass fractions																							
			SiO <sub>2</sub>		Al <sub>2</sub> O <sub>3</sub>		Fe <sub>2</sub> O <sub>3</sub>		CaO		K <sub>2</sub> O		MgO		Na <sub>2</sub> O		SO <sub>3</sub>		TiO <sub>2</sub>		P <sub>2</sub> O <sub>5</sub>		Other			
	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max		
1A	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
1B	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
1C	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
1D	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
1E	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
1F	0	3000	0.25	0.9	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.5	0	0.01	0	0.01	0	0.01	0	0.01		
2A	0	3000	0.4	0.65	0	0.3	0	0.2	0	0.2	0	0.2	0	0.2	0	0.2	0	1	0	0.01	0	0.01	0	0.01		
2B	0	3000	0.2	0.8	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	1	0	0.05	0	0.05	0	0.05		
2C	0	3000	0.2	0.8	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	1	0	0.05	0	0.05	0	0.05		
2D	0	3000	0.2	0.8	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	1	0	0.05	0	0.05	0	0.05		
2E	0	3000	0.2	0.8	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	0.4	0	1	0	0.05	0	0.05	0	0.05		
3A	0	3000	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		
3B	0	3000	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		
3C	0	3000	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		
3D	0	3000	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		
3E	1793	1913	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		
3F	1793	1913	0.1	0.6	0	0.3	0	0.03	0.1	0.6	0	0.03	0	0.3	0	0.03	0	1	0	0.03	0	0.03	0	0.03		

**Table 4.4: Selection criteria for each case study (part 2 of 2)**

Case Study	Gas atmosphere volume fractions																Viscosity (Pa·s)		Slag volume fraction		Excluded datasets	Exclude data without phase info?
	Ar		N <sub>2</sub>		O <sub>2</sub>		CO <sub>2</sub>		CO		H <sub>2</sub>		vaccum		unknown		Min	Max	Min	Max		
	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max	Min	Max						
1A	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	10 <sup>15</sup>	0	1	none	No
1B	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	10 <sup>15</sup>	0.01	1	none	Yes
1C	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	10	0	1	none	No
1D	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	10 <sup>3</sup>	10 <sup>6</sup>	0	1	none	No
1E	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	10	10 <sup>8</sup>	0	1	none	No
1F	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	10 <sup>8</sup>	10 <sup>15</sup>	0	1	none	No
2A	0	1	0	1	0	10 <sup>-7</sup>	0	1	0	1	0	1	0	1	0	0	1	100	0	1	none	No
2B	0	1	0	1	0	10 <sup>-7</sup>	0	1	0	1	0	1	0	1	0	0	1	100	0.01	1	none	Yes
2C	0	1	0	1	0	10 <sup>-7</sup>	0	1	0	1	0	1	0	1	0	0	1	100	0	1	none	No
2D	0	1	0	1	0	10 <sup>-7</sup>	0	1	0	1	0	1	0	1	0	0	0	10 <sup>15</sup>	0	1	none	No
2E	0	1	0	1	0	10 <sup>-7</sup>	0	1	0	1	0	1	0	1	0	0	100	10 <sup>5</sup>	0	1	none	No
3A	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	10 <sup>15</sup>	0	1	none	No
3B	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	1	0	10 <sup>15</sup>	0.01	1	none	Yes
3C	0.01	0.01	0.78	0.78	0.21	0.21	0	0	0	0	0	0	0	0	0	0	0	10 <sup>15</sup>	0	1	none	No
3D	0	0.01	0	0.78	0	0.21	0	0	0	0	0	0	0	0	1	0	10 <sup>15</sup>	0	1	none	No	
3E	0	0.01	0	0.78	0	0.21	0	0	0	0	0	0	0	0	1	0	10 <sup>15</sup>	0	1	none	No	
3F	0	0.01	0	0.78	0	0.21	0	0	0	0	0	0	0	0	1	0	10 <sup>15</sup>	0	1	Yakushev 1977	No	

**Table 4.5: Number of measurements identified and average absolute logarithmic error (AALE) of slag viscosity models for each case study**

Model	Case Study																
	1A	1B	1C	1D	1E	1F	2A	2B	2C	2D	2E	3A	3B	3C	3D	3E	3F
S2	1.372	1.161	0.761	N/A	2.509	5.109	0.192	0.256	0.266	1.749	0.465	0.694	0.677	0.213	0.772	1.782	0.165
Watt-Fereday	1.675	1.649	0.931	N/A	3.501	4.392	0.250	0.324	0.330	0.931	0.567	0.828	0.842	0.374	0.934	2.112	0.252
Bomkamp	3.763	3.886	3.124	N/A	5.891	3.843	1.579	1.431	1.435	2.072	1.141	1.684	1.718	1.536	1.863	2.346	1.760
Shaw	0.960	0.756	0.775	N/A	0.637	4.736	0.821	0.543	0.553	1.715	0.662	1.033	1.039	0.407	1.145	2.368	0.466
Lakatos	0.584	0.595	0.688	N/A	0.302	0.320	1.183	1.253	1.257	1.659	0.737	2.076	2.137	0.597	2.453	4.525	0.882
Urbain	0.766	0.443	0.253	N/A	1.072	6.523	0.651	0.487	0.490	1.649	0.376	0.591	0.564	0.181	0.636	1.439	0.169
Riboud	0.552	0.217	0.173	N/A	0.249	6.911	0.542	0.526	0.529	1.830	0.325	0.681	0.664	0.084	0.743	1.720	0.128
Streeter	1.735	1.145	0.990	N/A	1.685	12.006	0.618	0.522	0.531	1.526	0.401	0.981	0.969	0.263	1.095	1.580	0.213
Kalmanovitch-Frank	0.743	0.410	0.269	N/A	0.862	6.714	0.424	0.336	0.340	1.654	0.361	0.564	0.535	0.256	0.623	1.496	0.149
BBHLW	0.571	0.321	0.203	N/A	0.717	4.989	0.161	0.285	0.296	1.035	0.575	0.622	0.647	0.194	0.747	1.773	0.196
Duchesne	1.000	0.615	0.610	N/A	0.440	8.503	0.353	0.468	0.454	1.241	0.376	0.949	0.913	0.414	1.041	1.613	0.527
ANNliq	0.620	0.272	0.153	N/A	0.410	7.752	0.311	0.369	0.373	2.332	0.678	0.586	0.552	0.198	0.640	1.521	0.186
S2 +RE	N/A	1.170	0.777	N/A	2.524	N/A	0.195	N/A	0.279	N/A	0.418	N/A	0.691	N/A	N/A	1.769	0.165
Watt-Fereday +RE	N/A	1.680	0.949	N/A	3.514	N/A	0.245	N/A	0.325	N/A	0.522	N/A	0.842	N/A	N/A	2.078	0.251
Bomkamp +RE	N/A	3.922	3.142	N/A	5.912	N/A	1.622	N/A	1.473	N/A	1.278	N/A	1.775	N/A	N/A	2.323	1.761
Shaw +RE	N/A	0.794	0.793	N/A	0.677	N/A	0.696	N/A	0.518	N/A	0.586	N/A	1.087	N/A	N/A	2.342	0.464
Lakatos +RE	N/A	0.629	0.690	N/A	0.336	N/A	1.204	N/A	1.285	N/A	0.749	N/A	1.964	N/A	N/A	3.301	0.879
Urbain +RE	N/A	0.466	0.269	N/A	1.061	N/A	0.512	N/A	0.446	N/A	0.343	N/A	0.608	N/A	N/A	1.421	0.170
Riboud +RE	N/A	0.240	0.189	N/A	0.264	N/A	0.421	N/A	0.469	N/A	0.325	N/A	0.679	N/A	N/A	1.695	0.130
Streeter +RE	N/A	1.162	0.993	N/A	1.724	N/A	0.478	N/A	0.473	N/A	0.384	N/A	0.990	N/A	N/A	1.553	0.213
Kalmanovitch-Frank +RE	N/A	0.434	0.282	N/A	0.863	N/A	0.316	N/A	0.330	N/A	0.386	N/A	0.555	N/A	N/A	1.479	0.150
BBHLW +RE	N/A	0.335	0.219	N/A	0.664	N/A	0.191	N/A	0.319	N/A	0.522	N/A	0.685	N/A	N/A	1.749	0.195
Duchesne +RE	N/A	0.633	0.626	N/A	0.449	N/A	0.396	N/A	0.487	N/A	0.372	N/A	0.987	N/A	N/A	1.662	0.526
ANNliq +RE	N/A	0.282	0.167	N/A	0.410	N/A	0.293	N/A	0.368	N/A	0.595	N/A	0.571	N/A	N/A	1.506	0.186
Number of measurements	167	159	122	0	37	9	57	471	447	807	41	603	558	73	440	98	57

#### **4.5.1 Case Study 1 – Glass formation**

Many glasses are typically composed of SiO<sub>2</sub>-CaO-Na<sub>2</sub>O with a few trace elements. The glass melting process is described by Auchet et al. [22]. A glass viscosity below 10 Pa·s is required for refining and homogenizing, between 10<sup>3</sup> and 10<sup>6</sup> Pa·s for blowing and molding, and around 10<sup>12</sup> Pa·s for annealing [23]. In Case Study 1A, limits on the slag mass fractions were used to identify measurements with slags which are composed of 25-90 wt.% SiO<sub>2</sub>, 0-50 wt.% CaO, 0-50 wt.% Na<sub>2</sub>O and less than 1 wt% in other components. No limitations were put on the gas atmosphere composition since the oxidation states of SiO<sub>2</sub>, CaO and Na<sub>2</sub>O do not vary in liquid slag. In all, 167 measurements were identified. With these, the Riboud model is best (AALE 0.552). However, no results were obtained with the viscosity models applying the RE modification. This is due to selected measurements with a predicted liquid slag volume fraction of zero. In Case Study 1B, the measurements with a liquid slag volume fraction of zero were excluded. The Riboud model is still the best (AALE 0.217). Although the AALE is now lower, excluding measurements with a predicted liquid slag volume fraction of zero may remove measurements of interest. It is more important to note that the RE modification does not improve the model performances. Hence for the next glass formation cases, the limit on the liquid slag volume fraction was not used. In Case Study 1C, a model is desired specifically for the refining and homogenization process. Measurements above 10 Pa·s are excluded. 122 measurements were identified and the ANNliq model performs best (AALE 0.153). In Case Study 1D, a model is desired specifically for blowing and molding. Measurements are limited to 10<sup>3</sup>-10<sup>6</sup> Pa·s. In this case no data was identified. Therefore, the viscosity range was increased to 10-10<sup>8</sup> Pa·s in Case Study 1E. 37 measurements were identified and the Riboud model has the best performance (AALE 0.249). Case Study 1F is focused on the annealing process, with only measurements in the viscosity range 10<sup>8</sup>-10<sup>15</sup> Pa·s considered. 9 measurements are identified and the Lakatos model has the best performance (AALE 0.320). It should be noted that the Riboud model has an AALE of 6.911 in this case. Although the Riboud model has the best general performance for the glass compositions considered, other models should be applied when considering specific glass formation processes.

#### **4.5.2 Case Study 2 – Entrained flow gasification**

A thorough description of entrained flow gasification can be found in [24]. As a heuristic, slag viscosity should not exceed 25 Pa·s at the slag tap [9]. In Case Study 2A, mass fraction limits for the slag composition were specified in the SVD to identify measurements with compositions similar to typical coal ashes. A maximum volume fraction of  $10^{-7}$  was specified for O<sub>2</sub> in the gas atmosphere. Measurements with an unknown gas atmosphere were excluded. Although the fraction of O<sub>2</sub> is greater than  $10^{-7}$  in certain sections of the gasifier, it should be below this limit near the slag tap in a down-fired gasifier. It is also assumed that the slag in the gasifier will approach equilibrium with the gas. It is important to specify the gas atmosphere in this case since the viscosity of slags with high iron content will be significantly affected due to the multiple possible oxidation states of iron. Furthermore, since the heuristic limit is at 25 Pa·s, measurements in the range of 0-100 Pa·s were specified in the SVD. 57 measurements were identified and the BBHLW model has the best performance (AALE 0.161). In Case Study 2B, the composition limits were expanded to include more measurements. 471 measurements were identified and the S2 model has the best performance (AALE 0.256). Depending on the situation, more or less stringent selection criteria may be preferred, with more or fewer measurements available. The criteria used for Case Study 2B were kept for the following cases. To determine whether the RE modification can provide better model performances, measurements with a predicted liquid slag volume fraction of zero were excluded in Case Study 2C. The results show that the RE modification modestly worsened the performance of some models, and modestly improved that of others. Nonetheless, the S2 model still has the best performance (AALE 0.266).

For comprehensive CFD models of entrained flow gasifiers, slag viscosity predictions greater than 100 Pa·s are required to model slag flow along the reactor walls, and sometimes to predict the stickiness of slag particles. In Case Study 2D, the criteria for Case Study 2B were used, however no limit was placed on viscosity. In this case, the Watt-Fereday model has the best performance (AALE 0.931). An AALE of 0.931 translates to having predictions which are on average 8.53 times greater or smaller than the measured value. It may be more useful to evaluate the performance of models in different viscosity ranges. It was determined in Case Study 2B that the S2 model should be used in the 0-100 Pa·s range. It might be assumed that the slag flow in the gasifier is negligible above  $10^5$  Pa·s, and that the slag is no longer sticky above this same value. In Case Study 2E, viscosity was limited to the 100- $10^5$  Pa·s range. 41 measurements were identified and the Riboud model has the best performance (AALE 0.325).

However, by inspecting the selected measurement data, it should be noted that there are no measured viscosities greater than 625 Pa·s. Hence the Riboud model should be applied with great caution in the 625-10<sup>5</sup> Pa·s range.

### **4.5.3 Case Study 3 – Blast furnace**

The importance of slag viscosity in blast furnaces is described by Muller & Erwee [25]. In Case Study 3A, mass fraction limits for the slag composition were specified in the SVD to identify measurements with compositions similar to some typical blast furnace slags. 603 measurements are identified and the Kalmanovitch-Frank model has the best performance with an AALE of 0.564. To determine whether the RE modification can provide better model performances, measurements with a predicted liquid slag volume fraction of zero were excluded in Case Study 3B. The results show that the RE modification slightly decreased the performance of most models. The Kalmanovitch-Frank model still has the best performance (AALE 0.535). Phase limitations were not applied in the following case studies. Moreover, the blast furnace gas composition can vary throughout the furnace and with different operating conditions. In Case Study 3C, it is assumed that an oxidizing gas atmosphere is of interest. Only measurements performed in air were selected. 73 measurements were identified and the Riboud model has the best performance (AALE 0.084). However, more measurements may be desired to provide a better assessment of the models. Many of the measurements with an unknown gas atmosphere were likely performed in air. Therefore in Case Study 3D measurements in air or an unknown atmosphere were included. 440 measurements were identified and the Kalmanovitch-Frank model has the best performance (AALE 0.623). With more data available, other criteria may be specified to approach the conditions of interest. In Case Study 3E, the temperature range was limited to 1793-1913 K. 98 measurements were identified and the Urbain+RE model has the best performance (AALE 1.421). None of the models seem to have satisfactory performance in this case. Perhaps there is a specific dataset in the selected measurements which should not be trusted. In Case Study 3F, measurements from Yakushev 1977 were excluded. 57 measurements were identified and the Riboud model has the best performance (AALE 0.128).

## 4.6 Conclusion

A slag viscosity modeling toolbox, comprised of the SlagViscosityPredictor tool (SVP) and SlagViscosityDatabase tool (SVD), was developed to facilitate the evaluation and selection of slag viscosity models for given slag compositions and conditions. The functions of the SVP and SVD were presented. An artificial neural network for liquid slag viscosity prediction, ANNliq, was introduced. Furthermore, the Roscoe-Einstein (RE) model was applied in conjunction with viscosity models and FactSage phase predictions. Glass forming, entrained flow gasification and blast furnace case studies were presented to demonstrate how the SVD and SVP can be applied. Slag viscosity model assessment is limited by the quantity and quality of available slag viscosity measurement data. No one model is applicable to all slag compositions and conditions, and model performances vary greatly. Model assessment should be done with great care to balance criteria stringency and the measurement data quantity. With the case studies, it was found that the RE-modified viscosity models did not have the best performances. However, this may not be true in other cases. Also, the RE modification might provide better results with other viscosity models not included in the SVP and SVD, or with better phase predictions.

There is much room for improvement with the slag viscosity toolbox. More detailed data, particularly with shear rate provided, should be added to the database. Phase predictions are expected to improve as well. Certain applications will require much more measurement data with certain components, such as  $\text{Cr}_2\text{O}_3$  and  $\text{V}_2\text{O}_5$  for example. In addition, there are more slag viscosity models available, and some to be developed, which could increase prediction accuracy. Some of these models can account for non-Newtonian behavior and the effect of solid particle shapes and sizes. In the effort to improve the toolbox, external contributions are welcome. More sharing of viscosity data and models in formats which are easy to use is encouraged.

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## **Chapter 5. Flow behaviour of slags from coal and petroleum coke blends**

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## 5.1 Abstract

Petroleum coke may be used as a fuel for entrained-flow slagging gasification. It may be blended with coal to provide a more attractive feedstock. The coal provides the benefits of enhancing reactivity and increasing the amount of slag coating the gasifier walls, while the petroleum coke increases the heating value of the fuel blend. The slagging behavior of the petroleum coke or blend must be known to determine if it is a suitable feedstock. In the present study, the slag viscosities of coal, petroleum coke and coal/petroleum coke blends were measured in the temperature range of 1175-1650°C. Two different viscosity measurement apparatuses were used in separate laboratories. Some viscosity measurements were repeated to test reproducibility of the results. Also, slags with and without sulphur were tested to determine whether the effect of sulphur can be neglected. The largest deviation between repeated  $\log(\text{viscosity in Pa}\cdot\text{s})$  measurements is 0.11. Both petroleum coke slags tested are suitable for operation at or above 1275°C. The temperature range of operability for one coal slag tested increased when blended with petroleum coke ash, but did not significantly change for other coal slags tested. Some effects of blending on viscosity behaviour can be explained by network former and modifier theory. Other effects are attributed to solids formation which was investigated via slag quenching experiments and FactSage predictions. Below 1300°C, vanadium, a major component of petroleum coke ashes, promotes the formation of spinel which increases slag viscosity.

*Keywords:* Slag, Viscosity, Coal, Petroleum coke, Gasification

## 5.2 Introduction

Production of petroleum coke, a by-product of the oil refining industry, has been and is expected to continue to increase [1]. The major factors driving this increase include the rising demand for transport fuels, the use of heavier crude oils and new environmental regulations pushing for reduced waste and highly refined fuels. Due to petroleum coke's high heating value and low cost, there is much interest in its use as a primary fuel or in coal-petroleum coke blends. Although various technologies can utilize petroleum coke as a fuel for power generation, gasification can provide lower gaseous emissions and less solid waste [2]. Another advantage of gasification for petroleum coke conversion is the encapsulation of hazardous metals such as vanadium and nickel in non-leachable slag. Furthermore, gasification may be used to provide steam and hydrogen in addition to power generation. This can be beneficial when the gasifier is sited near the petroleum coke-producing oil refinery which requires steam, hydrogen and power [3].

The majority of the successful high throughput coal gasifiers developed in the past 60 years are of the entrained-flow slagging type [4]. In this type of gasifier, most of the inorganic component of the fuel (ash) is partially or fully melted, sticks to the reactor wall and flows to the bottom as slag. To determine the performance and profitability of a slagging gasifier it is important to know the slagging properties of the intended fuel. A slag which is too viscous may accumulate on the reactor wall until the reactor begins to plug, requiring corrective action which may include increasing the gasifier operating temperature, oxygen injection at the slag tap, or ceasing operation. As a rule of thumb, slag viscosity should not exceed 25 Pa·s at the slag tapping temperature [5,6]. If a fuel's slag is too viscous, the gasifier operating temperature may be increased, a fluxing agent such as limestone or dolomite may be added, and/or the fuel may be blended with another fuel. Blending fuels may also be necessary if there are issues with fuel availability, flame stability or ash content.

Of particular interest with petroleum coke slag is the effect of vanadium, a major component of the ash, on the viscosity of slags. Park *et al.* [7] obtained viscosity results for slags from a Korean anthracite blended with CaO and V<sub>2</sub>O<sub>3</sub>. V<sub>2</sub>O<sub>3</sub> content in the slags did not exceed 3.4 wt%. Park *et al.* focused on vanadium's effect on the temperature of critical viscosity; the temperature below which viscosity sharply increases upon further cooling. In the study of Park *et al.*, the temperature of critical viscosity increased slightly with increasing V<sub>2</sub>O<sub>3</sub> concentration. Vanadium's multiple oxidation states and complex phase formations [8,9] make it challenging to predict its interaction with the other components of slag.

In the present study, the viscosities of Australian and Canadian coals and petroleum coke slag blends were measured. Measurements using two apparatuses with different viscometer models, furnace models and operators were compared for reproducibility. Observed phases in quenched slags and predicted phases in these blends were analyzed to determine a correlation between solids formation and changes in viscosity. Additional viscosity and quenching tests were performed by adding pure vanadium oxide to coal ashes/slags in order to distinguish the effect of vanadium and other petroleum coke components on viscosity.

Accuracy of slag viscosity measurements are often questioned and commonly used slag viscosity model predictions may differ from measured values by orders of magnitude [10]. Past studies have compared results from various laboratories to demonstrate reproducibility of slag viscosity measurements [11,12]. Important sources of error include improper temperature measurement, crucible/spindle shape instability, shear-rate dependence of non-Newtonian samples, and compositional changes of the sample due to volatilization, reduction/oxidation by the gas atmosphere and/or reaction with the crucible/spindle. Measures to limit error, such as using calibrated temperature measurement devices, shear rate variation and post-measurement sample analyses have been undertaken here for measuring slag viscosity.

## 5.3 Experimental

### 5.3.1 Coal and petroleum coke selection

Several coal and petroleum coke ashes, slags, and representative synthetic oxide mixtures were used in this study. Compositions are given in Table 5.1. The listed compositions exclude some components which represent less than 1 wt.% of the ash and were normalized to 100%. Artificial coal ashes were prepared by mixing laboratory or analytical grade Al<sub>2</sub>O<sub>3</sub>, CaO, Fe<sub>2</sub>O<sub>3</sub>, K<sub>2</sub>CO<sub>3</sub>, MgO, Na<sub>2</sub>CO<sub>3</sub>, NiO, S, SiO<sub>2</sub>, TiO<sub>2</sub> and V<sub>2</sub>O<sub>5</sub> powders. Real coal and petroleum coke ashes were prepared by ashing of the coal or petroleum coke at 700-750 °C for 20-30 h. Slags S1 and S2 were collected from Siemens' pilot plant gasifier in Germany. Slags S3 and S4 were collected from CanmetENERGY's pilot plant gasifier in Canada.

**Table 5.1: Composition of coal and petcoke ashes/slags**

Oxide	Weight%								
	P1a	P1b	P2	C1a	C1b	S1	S2	S3	S4
SiO <sub>2</sub>	37.77	40.66	43.41	58.02	60.01	49.62	47.73	49.24	46.54
Al <sub>2</sub> O <sub>3</sub>	16.60	17.87	14.82	23.47	24.27	18.22	24.07	21.02	21.29
Fe <sub>2</sub> O <sub>3</sub>	7.35	7.91	9.41	4.55	4.71	24.36	15.41	7.38	12.21
TiO <sub>2</sub>	1.10	1.19	0.62	0.53	0.54	1.17	1.49	0.68	1.02
CaO	15.82	17.03	4.13	5.63	5.82	3.27	8.25	13.15	10.34
MgO	3.87	4.17	1.36	1.57	1.62	1.62	0.97	2.85	2.47
Na <sub>2</sub> O	1.67	1.80	1.13	2.34	2.42	0.55	0.28	2.71	1.43
K <sub>2</sub> O	0.96	1.03	1.75	0.58	0.60	0.45	0.30	0.82	0.67
V <sub>2</sub> O <sub>5</sub>	6.67	7.18	20.69	<1	0.01	<1	<1	<1	<1
NiO	1.09	1.17	2.68	<1	<1	<1	<1	<1	0.61
Cr <sub>2</sub> O <sub>3</sub>	<1	<1	<1	<1	<1	<1	<1	2.15	3.43
P <sub>2</sub> O <sub>5</sub>	<1	<1	<1	<1	<1	0.73	1.49	<1	<1
SO <sub>3</sub>	7.11	<1	<1	3.32	<1	<1	<1	<1	<1

P1a is real ash produced from an Alberta, Canada oil sands delayed coke. P1b is an artificial ash based on the composition of P1a, excluding sulphur. P2 is an artificial ash based on the ash composition of petroleum coke sourced from an eastern Canadian power utility, excluding sulphur. C1a is an artificial ash based on the ash composition of an Alberta, Canada sub-bituminous coal. C1b is the equivalent of C1a without sulphur. S1 is slag produced from gasification of a sub-bituminous coal from Western Australia, Australia. S2 is slag produced from gasification of a high volatile bituminous coal from New South Wales, Australia. S3 is slag produced from the gasification of a sub-bituminous coal from Alberta, Canada. S4 is slag produced from the gasification of a 50/50 blend of S3 and P2's parent fuels.

Although S4 is from a 50/50 blend of S3 and P2's parent fuels, its composition can not be calculated based on the compositions of S3 and P2. Based on the compositions of S3 and P2, S4 is expected to contain several weight percent  $V_2O_5$ . However, S4 does not contain an appreciable amount of  $V_2O_5$ . It is believed that due to lower reactivity, the petroleum coke ash reported to the fly ash stream rather than the coarse slag. Also, S3 and S4 have higher  $Al_2O_3$  and  $Cr_2O_3$  than the ash of their parent fuels. These components are believed to have been taken up by the slag from the gasifier refractory. Future testing will be done to study enrichment and depletion in the gasifier of various slag components.

The base coal and petroleum coke ashes/slugs, as described above, were blended for viscosity measurements and quenching experiments. For sample identification in this paper, the identifier of the majority ash is listed first, followed by the wt.% (relative to the majority ash) and identifier of the secondary ash. For example, C1b+25P1b contains 25 g of P1b for every 100 g of C1b.

### **5.3.2 Viscosity measurements**

Slag viscosity measurements at CSIRO, Australia, were performed using a Haake high temperature viscometer using a rotational bob [13]. Ashes were placed in a molybdenum crucible and heated to 1200 °C and left to equilibrate overnight. They were then heated to 1520-1550 °C in a nitrogen atmosphere. Sample temperature was measured with a

thermocouple placed near the sample location. The crucible was placed in a sacrificial graphite sleeve to consume free oxygen in the constant nitrogen gas flow. Oxygen partial pressure ( $p_{O_2}$ ) in the system has been measured using an oxygen probe (HT oxygen probe, Australian Oxytrol Systems). At temperatures between 1200-1450 °C,  $p_{O_2}$  values near the slag surface were between  $2.8 \times 10^{-6}$  and  $2.9 \times 10^{-4}$  Pa, confirming the reducing nature of the experimental conditions. Viscosity measurements were performed using a molybdenum rotating bob connected to a viscometer head. Measurements were conducted at incremental temperature reduction steps within the temperature range 1200-1550 °C. The cooling step for each set of measurements was of 25 °C with at least 30 min intervals to equilibrate temperatures and compositions inside the crucible. At each step, shear stress is measured with varying shear rate to detect non-Newtonian behaviour, and the slag viscosity was measured several times to ensure the repeatability of measurements. The average of these measurements was used as the value of viscosity [14].

Slag viscosity measurements at CanmetENERGY, Canada, were performed in a Carbolite 3 litre BLF1700 furnace with a Brookfield RVDV-IIIU rheometer. Both are connected to a PC for external control. Unless otherwise indicated, samples are placed in a molybdenum crucible. The crucible is placed in a custom-built alumina chamber within the furnace for gas atmosphere control and spill protection. A type B thermocouple is inserted into the chamber to monitor the sample temperature. Argon is fed at 0.4 dm<sup>3</sup>/min and a graphite sleeve is placed around the molybdenum crucible to consume trace oxygen. Mössbauer spectroscopy of a slag sample after viscosity measurements indicates that all iron atoms are in the ferrous state, confirming the reducing nature of the experimental conditions. A molybdenum spindle, attached to the rheometer, reaches the slag sample and is rotated to measure torque. The operation of the rheometer was validated with the National Institute of Standards and Technology's (NIST, United States) standard reference material 717a. Ash samples were first heated to 1225 °C and left overnight for equilibration. They were then heated to 1475-1650 °C. Measurements were conducted at incremental temperature reduction steps within the temperature range 1175-1650 °C. The cooling step for each set of measurements was of 25 °C with at least 25 min intervals to equilibrate temperatures and compositions inside the crucible. At each

temperature, viscosity was measured multiple times at various rotational speeds to detect non-Newtonian behaviour. The final measured value is taken at approximately 80-90% of the rheometer's maximum torque.

### **5.3.3 Quenching experiments**

Two techniques were used to quench slag samples. The first one was cold rod quenching which involved dipping of a molybdenum rod in the molten slag after viscosity measurements, removing the rod and then cooling it in water. The second one was quenching in a vertical tube furnace. For this, a sample of slag or a pelletised ash mixture was placed in a molybdenum envelope and suspended on a molybdenum wire in the cold zone at the bottom of the vertical tube furnace. The lower end of the work tube was sealed with a plastic film and submerged in water. The furnace tube was flushed with nitrogen during all experiments and graphite was attached to the molybdenum envelope to ensure reducing conditions similar to those used in the viscosity measurements. The sample was raised to the hot zone of the furnace and equilibrated at the selected temperature for 5–15 h and then quenched by dropping into the water.

### **5.3.4 Sample analysis**

Bulk major and minor oxide concentrations in samples were determined by x-ray fluorescence. Quenched slags and slags remaining after viscosity measurements were examined using Scanning Electron Microscopy (SEM) in backscattering mode. Energy Dispersive X-ray Spectroscopy (EDS) was undertaken to identify the elemental composition of selected regions of the slag. Electron Probe Microanalysis (EPMA) was used to identify the compositions of solid and liquid phases of quenched samples.

### **5.3.5 FactSage predictions**

FactSage software predicts equilibrium solid-liquid-gas phases and compositions based on Gibbs free energy minimization [15]. Gibbs free energy is calculated from optimized models with parameters based on empirical data with various compositions, temperatures

and pressures. This information is contained within compound and solution databases which must be carefully selected prior to equilibrium calculations. The FactSage 6.2 Equilib module was utilized with the FACT53 and FToxid databases. 1 g of N<sub>2</sub> per 100 g of sample was included in the calculations. Some oxide components representing less than 1 wt.% of the samples were excluded. Of all gas compounds, solid compounds and solutions which can be formed, only solutions containing titanium were excluded. These components and solutions were excluded as to not exceed the maximum phase limits in FactSage.

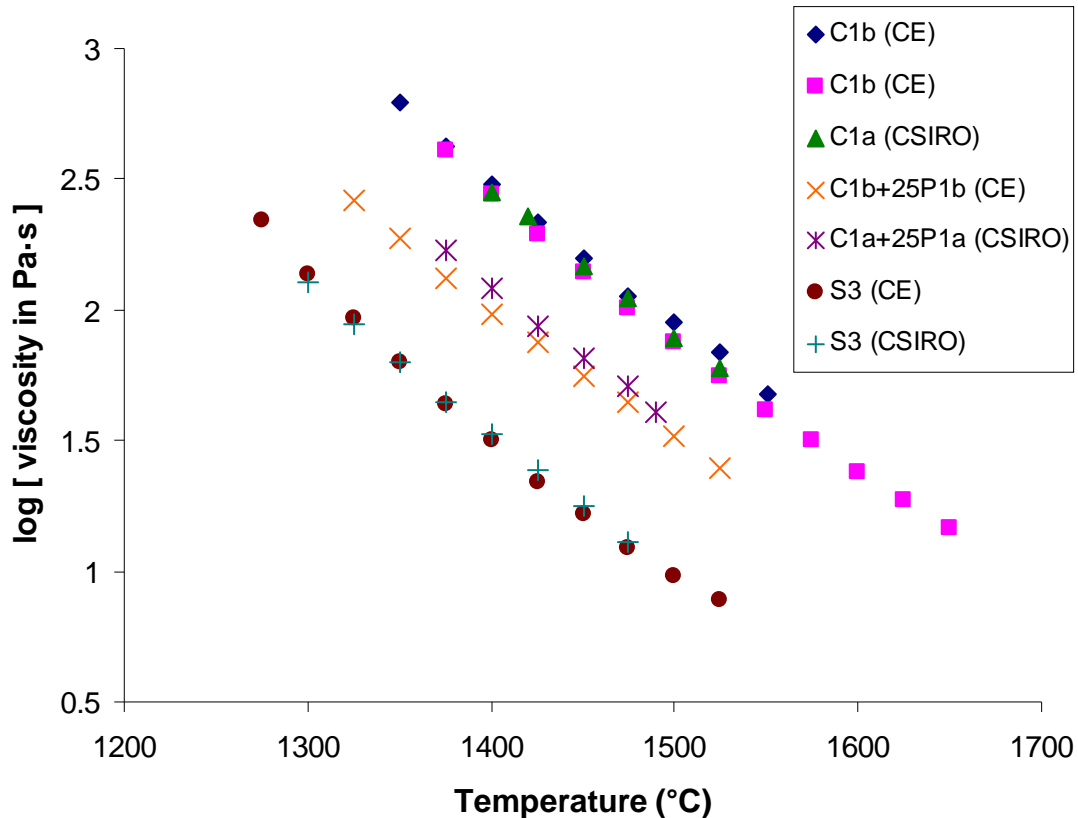
## 5.4 Results

### 5.4.1 Validation of viscosity measurements

Viscosity values are given as logarithmic (base 10) values, with viscosity in Pa·s. This makes it easier to view trends in temperature-viscosity curves since a large range of viscosity is covered and the logarithm viscosity, herein referred to as LV, is roughly a linear function of temperature. Furthermore, differences between LVs can be interpreted as a factor difference between viscosities. For example, LVs with a difference of 0.2 correspond to viscosities with a factor difference of 1.58 (i.e.,  $10^{0.2}$ ).

Various temperature-LV curves were generated to test the repeatability of results with the CanmetENERGY viscosity measuring apparatus, and the reproducibility of results between the CanmetENERGY and CSIRO viscosity apparatuses (Figure 5.1). Two temperature-LV curves were generated four months apart for C1b as a measure of repeatability with the CanmetENERGY viscosity equipment. Over the temperature range of 1400-1550 °C, and the LV range of 1.62-2.48, the greatest deviation in LV is 0.09. Reproducibility between the CanmetENERGY and CSIRO viscosity rigs was determined by measuring the viscosity of S3 with both. Over the temperature range of 1300-1475 °C, and the logarithmic viscosity range of 1.09-2.14, the greatest deviation in LV is 0.04. A few viscosity measurements were conducted to determine whether sulphur can be

excluded from artificial samples without affecting viscosity since most of the sulphur volatilizes from quenched slag samples which have been above 1200 °C. These measurements also served as additional tests of reproducibility. C1b is the same as C1a without sulphur. C1b+25P1b is the same as C1a+25P1a without sulphur and a few minor components. Greater LV differences were obtained between C1b+25P1b (CanmetENERGY) and C1a+25P1a (CSIRO). The largest difference is 0.11 at 1375 °C.



**Figure 5.1: Temperature-LV curves demonstrating repeatability and reproducibility of slag viscosity measurements, with and without sulphur, at CanmetENERGY (labeled CE) and CSIRO (labeled CSIRO).**

#### **5.4.2 Viscosity of blends**

Temperature-LV curves for coal and petroleum coke slag blends are given in Figure 5.2. Dashed lines are drawn at 25 Pa·s, which is considered the upper limit of operability. For C1b and P1b blends, increasing the P1b content reduces viscosity over the entire

temperature range tested, with pure P1b having the lowest viscosity. C1b has a viscosity above 25 Pa·s for the entire temperature range tested, while C1b+100P1b is suitable for operation above 1400 °C. S1 and its blends all cross the 25 Pa·s line between 1270 and 1290 °C. At 1450 °C, S1+P1a blends have LV's which are up to 0.16 lower than the LV of S1. At 1200 °C, S1+P1a blends have LV's which are up to 0.21 higher than the LV of S1. Viscosity of pure P1b, assumed to be the same as P1a, is lower over the entire temperature range tested, albeit with a sharper increase in viscosity upon cooling. S2 and its blends all cross the 25 Pa·s line between 1310 and 1340 °C. At 1475 °C, S2+50P1a has an LV which is 0.08 lower than the LV of S2. At 1250 °C, S1+50P1a has an LV which is 0.07 higher than the LV of S2. Viscosity of pure P1b is lower over the entire temperature range tested, albeit with a sharper increase in viscosity upon cooling. When S2 was mixed with 2.2 wt% V<sub>2</sub>O<sub>5</sub>, which is the same amount of V<sub>2</sub>O<sub>5</sub> as in S2+50P1a, the LV was higher than S2's LV by 0.04 at 1450 °C, and by 0.21 at 1300 °C. Both S3 and S4 have a viscosity below 25 Pa·s at temperatures above 1400 °C. S4 has a lower viscosity than S3 at temperatures above 1400 °C, and a higher viscosity at temperatures below 1400 °C. Adding 5 wt% V<sub>2</sub>O<sub>5</sub> to S3 reduces the LV by 0.28 at 1525 °C, but increases the LV by 0.17 at 1300 °C. P2, based on the petroleum coke which had been used to produce S4, has a viscosity below 25 Pa·s over the entire temperature range tested. Note that the viscosity of P2 was measured with an alumina crucible and spindle. An attempt had been made to measure its viscosity with a molybdenum crucible and spindle; however, the viscosity did not stabilize after 4 hours at 1475 °C.

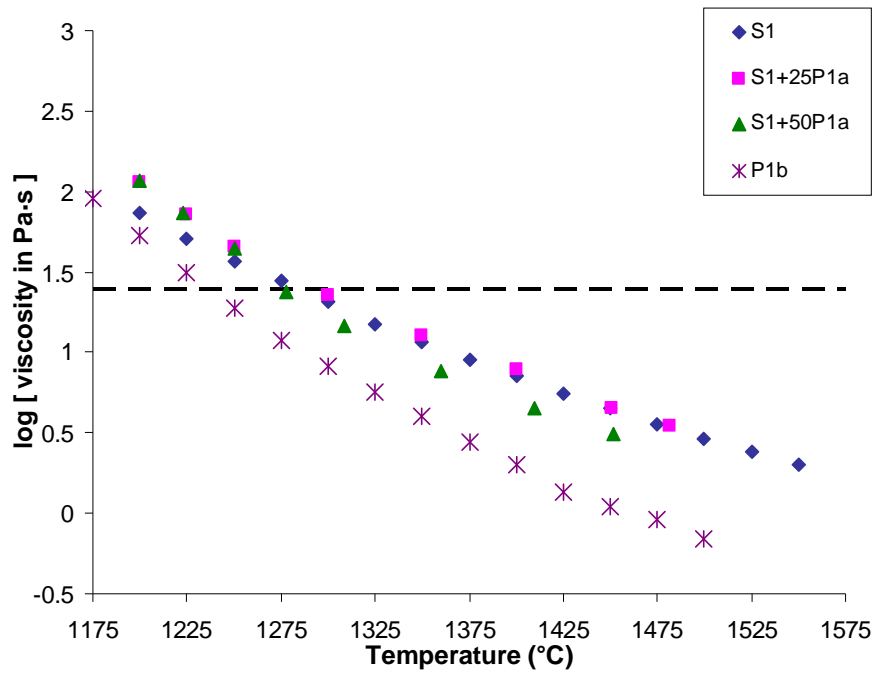
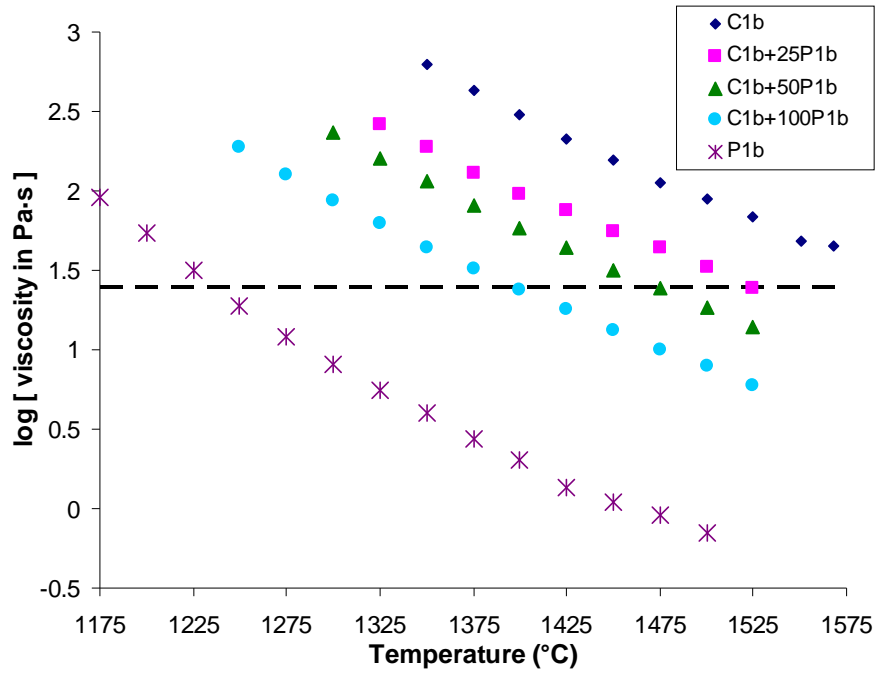


Figure 5.2: Temperature-LV curves of coal and petcoke slag blends. Dashed lines are drawn at 25 Pa·s. (part 1 of 2)

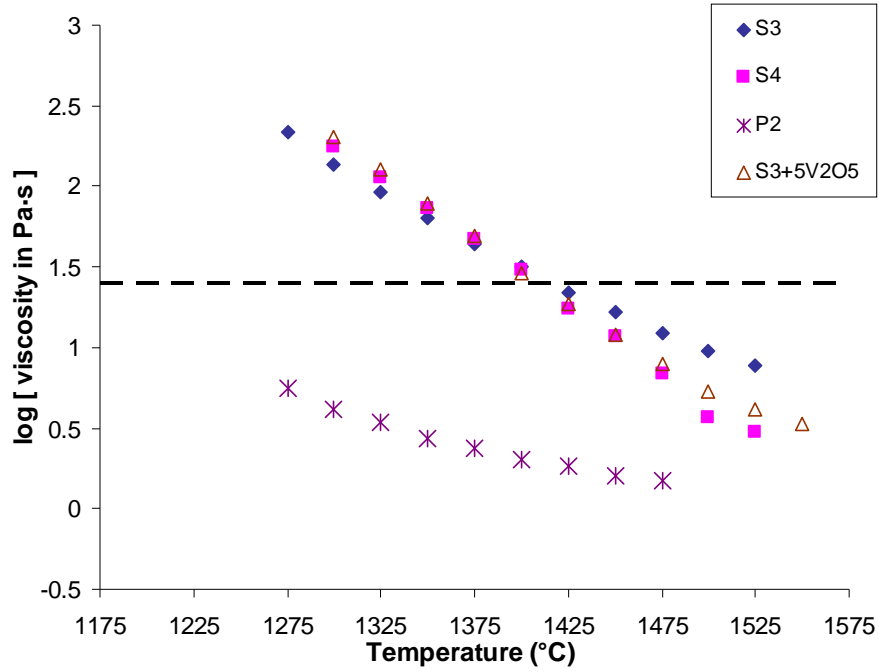


Figure 5.2: Temperature-LV curves of coal and petcoke slag blends. Dashed lines are drawn at 25 Pa·s. (part 2 of 2)

### 5.4.3 Slag microstructure

In order to understand the viscosity behaviour of investigated slags below the liquidus temperature, a series of slag quenching experiments was conducted. Quenching temperatures were selected near the lower temperature limit of the viscosity measurements. Figure 5.3 shows micrographs of the C1b and C1b+xP1b slags quenched from 1350 °C and 1300 °C. At 1350 °C, only C1b slag contains solids. These are identified by EPMA as mullite. Both blended samples have only a liquid phase at this temperature. At 1300 °C, C1b slag still has more solids than C1b+xP1b blends. In the blends, vanadium-rich spinels are present. It is clear that for C1b coal ash, blending with petroleum coke ash significantly changes phase equilibria in the slag system and lowers the liquidus temperature. Figure 5.4 shows micrographs of the S1 and S1+xP1a slags quenched from 1200 °C. S1 slag is fully liquid, while S1+25P1a and S1+50P1a blends have spinel and spinel with feldspar, respectively. Figure 5.5 shows micrographs of the S2, S2+50P1a and S2+2.2V<sub>2</sub>O<sub>5</sub> slags quenched from 1250 °C. The pure S2 slag has only a small amount of mullite. For S2+50P1a slag, the composition shifts to the spinel primary phase field. For S2 mixed with 2.2 wt% V<sub>2</sub>O<sub>5</sub>, which is the same amount of V<sub>2</sub>O<sub>5</sub> as in S2+50P1a, feldspar and vanadium-rich spinel are present. Figure 5.6 shows micrographs of the S3, S4 and S3+5V<sub>2</sub>O<sub>5</sub> slags quenched from 1325 °C and 1275 °C. At 1325 °C, S3 has chromium-rich spinel and feldspar, while S4 has chromium-rich spinel containing a small amount of vanadium oxide (~ 2 wt.%). At 1275 °C, both S3 and S4 have chromium-rich spinel and a significant amount of feldspar. The chromium-rich spinel in S4 also contains some (~2 wt.%) vanadium oxide. For S3 mixed with 5 wt.% V<sub>2</sub>O<sub>5</sub>, which is in the same proportion as for the parent fuels of S4, chromium-vanadium spinel is present at 1325 °C. At 1275 °C, feldspar appears as a second solid phase. The quenched S3+5V<sub>2</sub>O<sub>5</sub> samples are very similar to the quenched S4 samples.

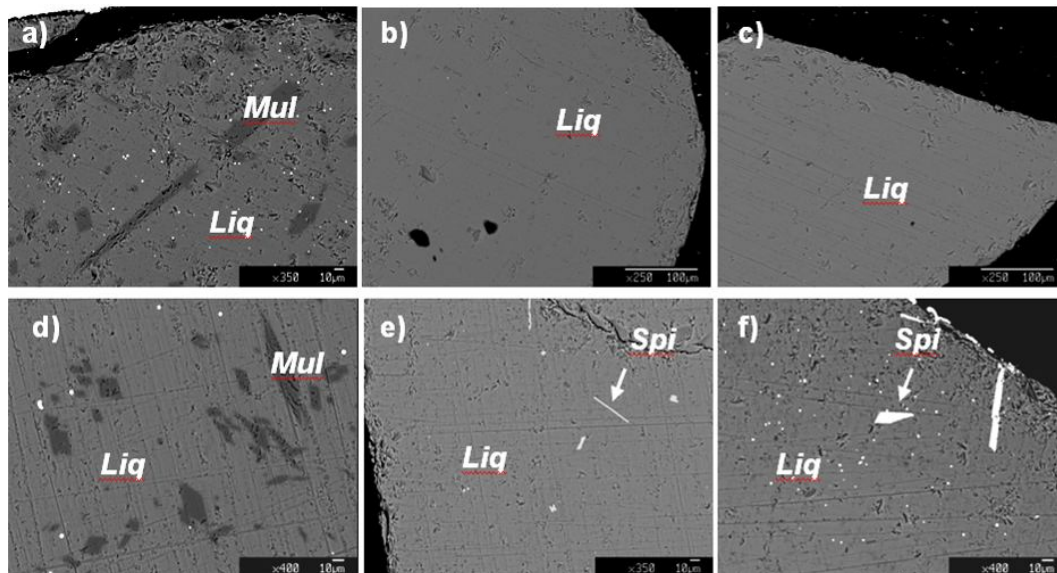


Figure 5.3: Microstructure of slags quenched from 1350 °C: C1b (a), C1b+50P1b (b), C1b+100P1b (c); and quenched from 1300 °C: C1b (d), C1b+50P1b (e), C1b+100P1b (f). Legend: Liq- former liquid phase, Mul- mullite, Spi- spinel.

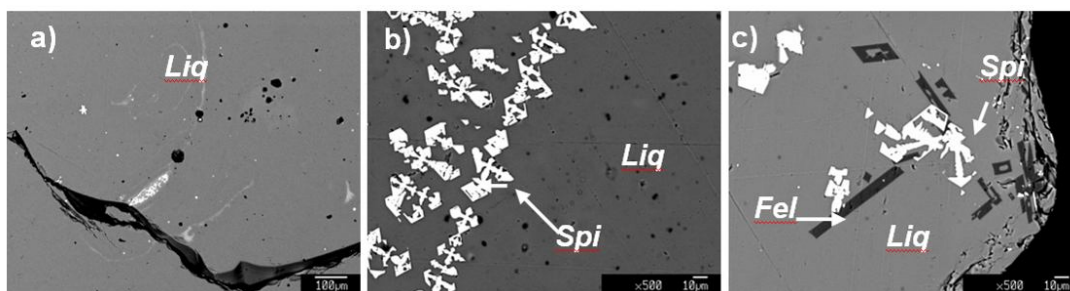


Figure 5.4: Microstructure of slags quenched from 1200 °C: S1 (a), S1+25P1a (b), S1+50P1a (c). Legend: Liq- former liquid phase, Fel- feldspar, Spi- spinel.

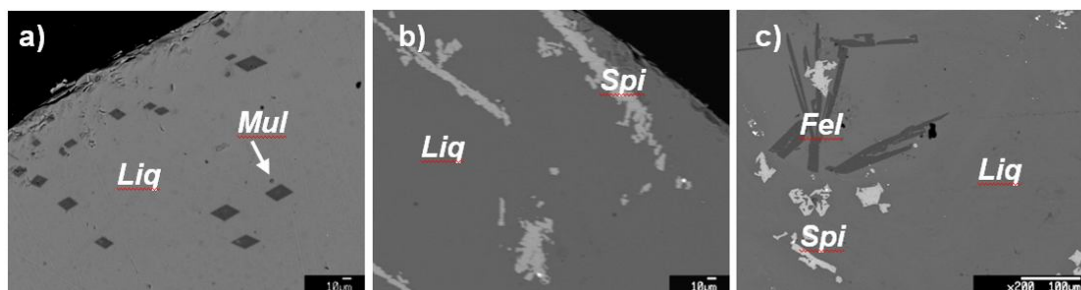
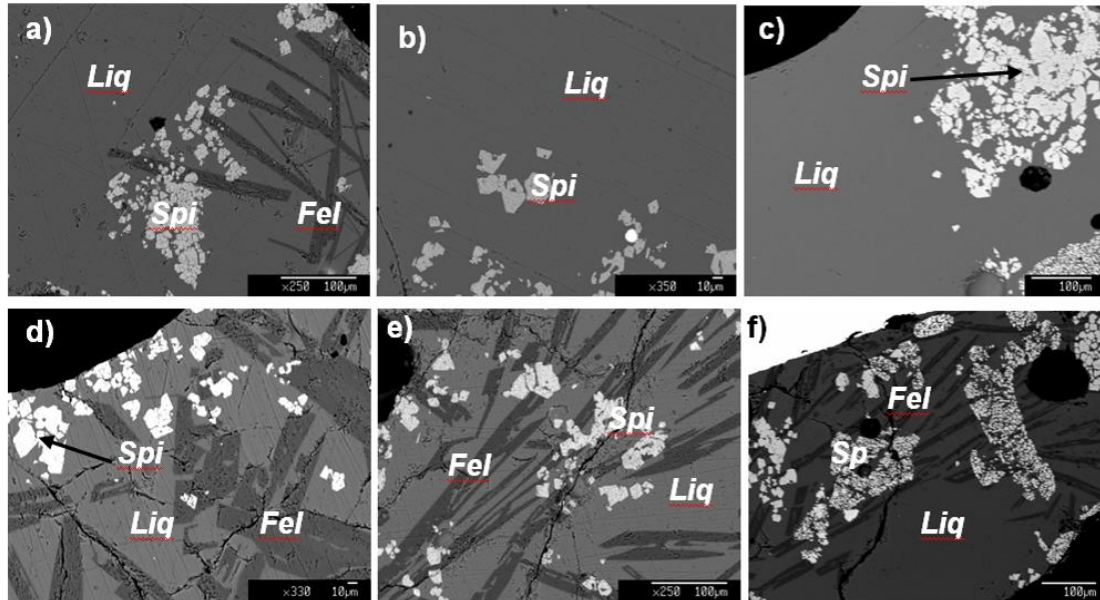


Figure 5.5: Microstructure of slags quenched from 1250 °C: S2 (a), S2+50P1 (b), S2+2.2V<sub>2</sub>O<sub>5</sub> (c). Legend: Liq- former liquid phase, Fel- feldspar, Mul- mullite, Spi- spinel.



**Figure 5.6: Microstructure of slags quenched from 1325 °C: S3 (a), S4 (b), S3+5V<sub>2</sub>O<sub>5</sub> (c); and quenched from 1275 °C: S3 (d), S4 (e), S3+5V<sub>2</sub>O<sub>5</sub> (f). Legend: Liq- former liquid phase, Fel- feldspar, Spi- spinel.**

The mass fractions of solids in the coal and petroleum coke slag blends, as predicted by FactSage, are given in Figure 5.7. The amount and type of solids predicted (type data not shown) generally do not match with results from quenching experiments. For instance, below 1300 °C, the FactSage predicted solids fractions are above 20% for most samples tested. Also, since vanadium is only included as solid compounds and not as slag solution species in the FactSage databases, all vanadium is predicted to be in solid form for the entire temperature range of all samples which contain vanadium.

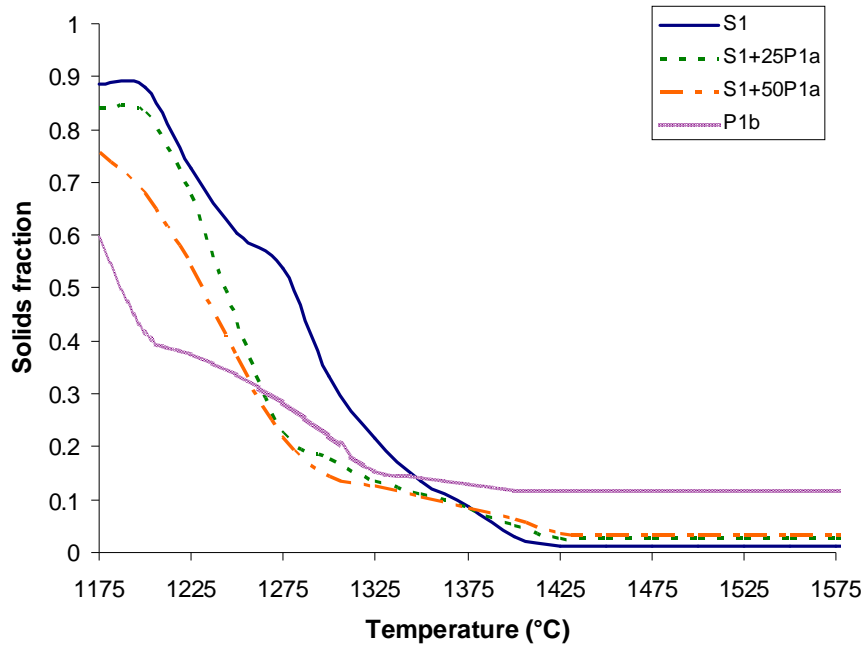
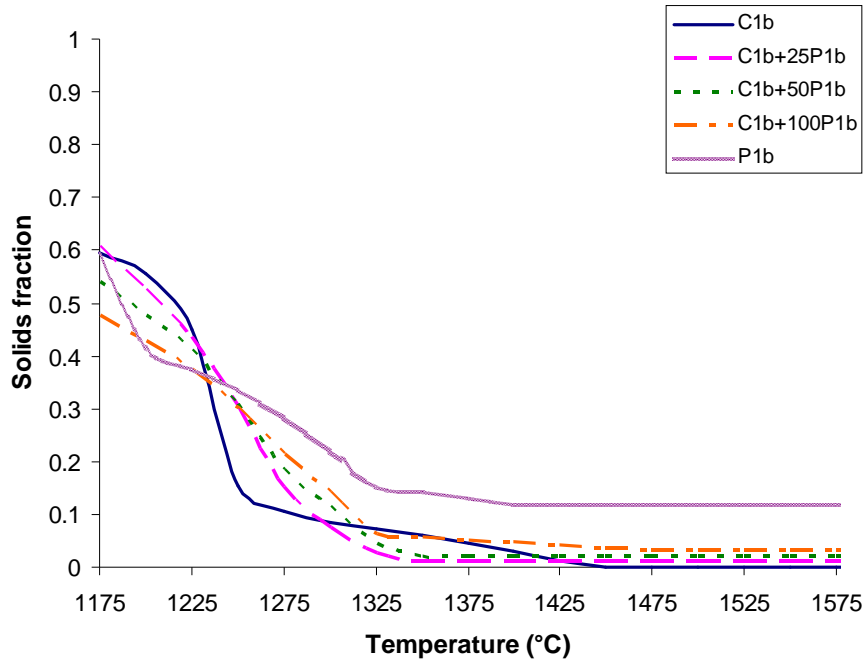


Figure 5.7: Predicted solids mass fractions of coal and petcoke slag blends using FactSage. (part 1 of 2)

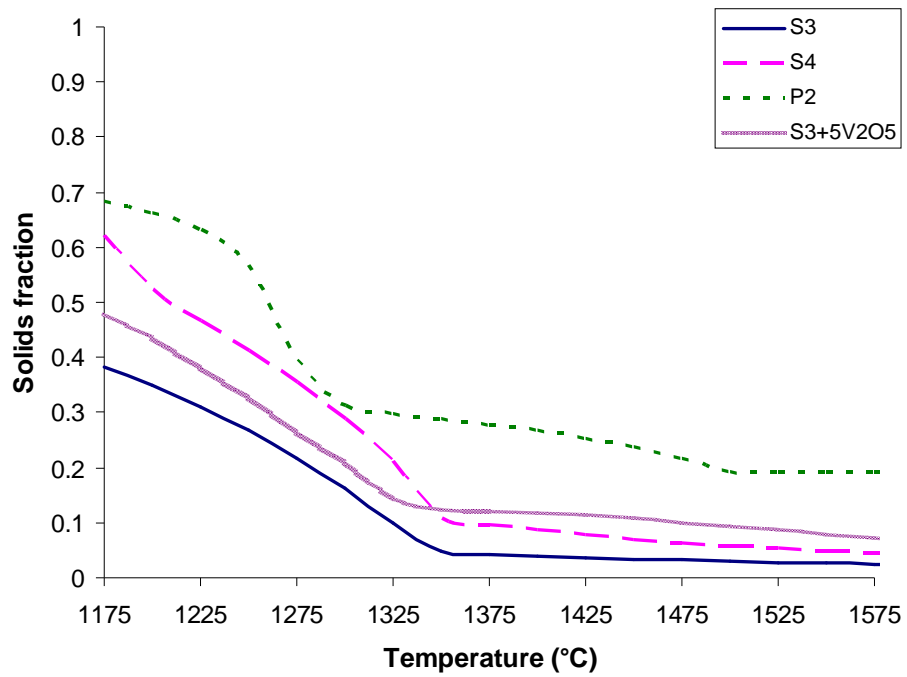
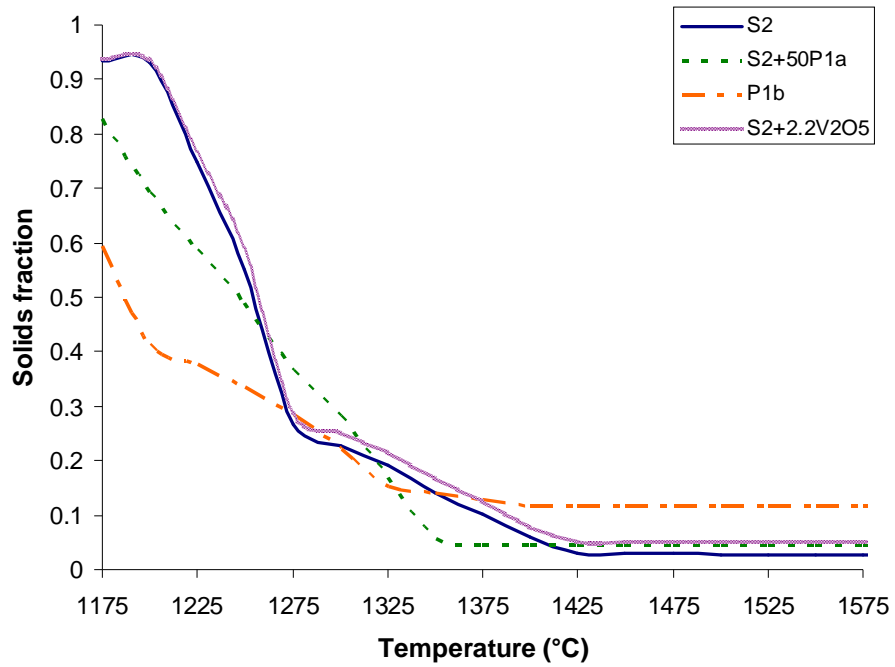


Figure 5.7: Predicted solids mass fractions of coal and petcoke slag blends using FactSage. (part 2 of 2)

#### 5.4.4 Reactivity with crucible and spindle

Post-run analysis of the slags indicates that  $\text{MoO}_3$  content does not exceed 1 wt.% for slags containing  $<5$  wt.%  $\text{V}_2\text{O}_5$ . For these slags, 10-15% of the  $\text{V}_2\text{O}_5$  in the slag is lost by reaction with the molybdenum crucible. Changes in composition are greater for slags with  $\geq 5$  wt.%  $\text{V}_2\text{O}_5$ . Figure 5.8 shows microstructures of P2 after viscosity tests conducted in both molybdenum and alumina crucibles. Slag-crucible reaction products, indicated by arrows, are observed at the slag-crucible interface. EPMA analysis reveals that the reaction products are vanadium-molybdenum metal and vanadium-containing spinel for the viscosity measurements carried out in molybdenum and alumina crucibles, respectively. Table 5.2 shows bulk compositions of vanadium-rich slags before and after viscosity measurements. For slags processed in molybdenum crucibles, the  $\text{V}_2\text{O}_5$  content decreases by 3-10 percentage points by mass during viscosity tests. This is equivalent to a loss of up to 66% of the initial  $\text{V}_2\text{O}_5$  content.  $\text{MoO}_3$  content is increased by up to 10 wt%. Compositional changes of P2 processed in an alumina crucible include a loss of 50% of the initial  $\text{V}_2\text{O}_5$  content, and an increase by 6 wt.% of the  $\text{Al}_2\text{O}_3$  content.

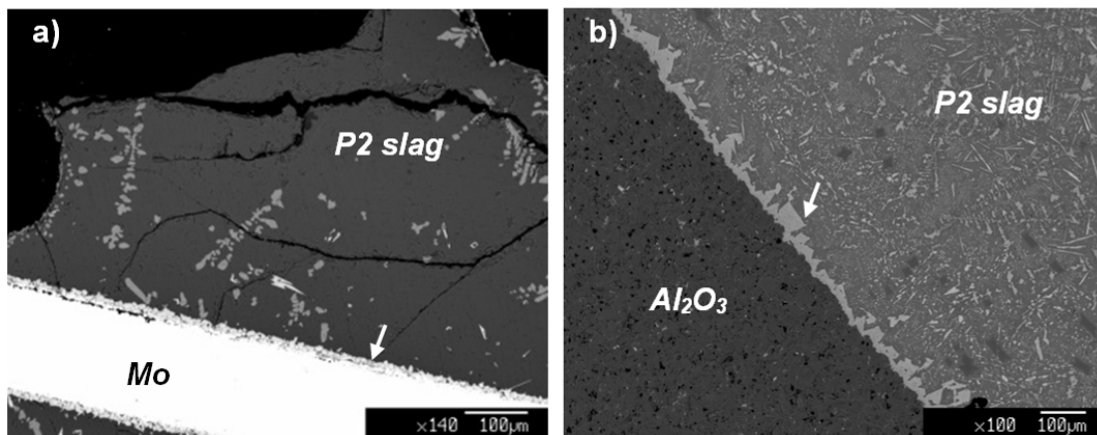


Figure 5.8: Microstructure of slags in Mo (a) and alumina (b) crucibles.

**Table 5.2: Composition of vanadium-rich slags before and after viscosity tests (key differences are in red fonts)**

	S3+10V <sub>2</sub> O <sub>5</sub>	S3+10V <sub>2</sub> O <sub>5</sub> in Mo	P2	P2 in Mo	P2 in Al <sub>2</sub> O <sub>3</sub>	P1b	P1b in Mo
SiO <sub>2</sub>	45.37	48.69	43.41	42.28	47.49	40.66	41.25
Al <sub>2</sub> O <sub>3</sub>	19.37	19.14	14.82	13.82	20.05	17.87	17.94
Fe <sub>2</sub> O <sub>3</sub>	6.8	7.18	9.41	9.56	7.23	7.91	7.53
TiO <sub>2</sub>	0.63	0.54	0.62	0.52	0.32	1.19	1.03
P <sub>2</sub> O <sub>5</sub>	0	0.5	0	0.1	0.04	0	0.05
CaO	12.12	12.48	4.13	3.72	3.52	17.03	16.03
MgO	2.63	2.43	1.36	1.13	1.73	4.87	3.65
Na <sub>2</sub> O	2.5	1.86	1.13	1.12	1.06	1.8	1.81
K <sub>2</sub> O	0.76	1.02	1.75	1.67	1.74	1.03	0.97
MoO <sub>3</sub>	0	2.3	0	10.64	0	0	5.16
V <sub>2</sub> O <sub>5</sub>	9.21	3.13	20.69	13.46	10.17	7.18	4.13
NiO	0	0.11	2.68	1.94	1.91	1.17	0.42

## 5.5 Discussion

Based on tests of repeatability and reproducibility, the largest deviation between LV datasets is 0.11. This deviation includes error induced by neglecting the effect of sulphur. Hence in the following, both C1a and C1b will be referred to as C1. Both P1a and P1b will be referred to as P1. As for compositional changes during viscosity measurements, some dissolution of molybdenum in coal slags was previously reported by French *et al.* [16]. In the present study, there was little contamination from molybdenum for slags with <5 wt.% V<sub>2</sub>O<sub>5</sub>. Only 10-15% of the V<sub>2</sub>O<sub>5</sub> was lost from the bulk slag. For slags with ≥5 wt.% V<sub>2</sub>O<sub>5</sub>, there was significant reaction with the molybdenum crucibles. As an alternative to molybdenum, alumina crucibles and spindles may be used for viscosity measurements. However, significant compositional changes were also observed for P2 processed in an alumina crucible. More investigations are required to find the optimum materials for viscosity measurements of vanadium-rich slags.

Differences in slag viscosities can be related to the slag compositions. Liquid slag components can be classified into three categories; network formers, network modifiers and amphoteric [10]. Network formers (e.g., Si) increase viscosity. Network modifiers (e.g., Na, K, Mg, Ca) decrease viscosity. Amphoteric (e.g., Al, Fe) can act either as network formers or modifiers. Other components have not been as thoroughly studied and are harder to classify (e.g., Ti, V, Ni, Cr). A previous study produced viscosity data for blends of C1, P1, limestone and/or dolomite [17]. It was shown that both limestone and dolomite are effective for reducing the viscosity of C1 and C1/P1 slags. The addition of 25 wt.% P1 also reduces the viscosity of C1, but not to the same extent as adding 25 wt.% limestone or dolomite. In the current study, the effect of adding petroleum coke ash to C1, S1, S2 and S3 was investigated. Based on the network former and modifier contents, adding P1 to C1, S1 and S2 should decrease their viscosity. Also, S4 should have a lower viscosity than S3. From the results, adding P1 to C1 decreases viscosity over the entire temperature range tested. Adding P1 to S1 and S2 only decreases the viscosity at certain temperatures. S4's viscosity is only lower than S3's viscosity above 1375°C. The effect of amphoteric and unclassified components may be causing the viscosity to increase. Another factor which causes viscosity to increase is the presence of solids.

Based on quenched sample analyses, addition of P1 to C1 reduced the liquidus temperature which, in addition to the network former and modifier contents, contributed to the lower viscosities of the C1+xP1 blends compared to C1. Adding P1 to S1 increased the liquidus temperature, which led to an increase in viscosity below 1275 °C. For S2, S2+50P1 and S2+2.2V<sub>2</sub>O<sub>5</sub>, a difference in liquidus temperature was not detected by quenching experiments. This is also true for S3, S4 and S3+5V<sub>2</sub>O<sub>5</sub>. However, more quenching experiments may reveal a difference in liquidus temperature. For all slags which contain vanadium, spinel is present in slags quenched at or below 1300 °C. This is likely associated to the sharper increase in viscosity upon cooling of these slags, compared to the slags which do not contain vanadium. FactSage results are ambiguous since the solid fraction curves are similar and crossover multiple times, particularly if one

does not consider vanadium solids at higher temperatures. Also, FactSage predictions have very high solids fractions for most slags below 1300 °C. Given the quenched sample observations, this has been shown not to occur.

C1 has a very high viscosity that makes its parent fuel a poor candidate as a feedstock for slagging gasification. Adding P1 to C1 reduced the slag viscosity to below 25 Pa·s over part of the temperature range tested. When P1 is added to S1 or S2, the temperature at which the slag viscosity reaches 25 Pa·s varies by less than 30 °C. S3 and S4 both cross the 25 Pa·s threshold at nearly the same temperature. At and above 1275 °C, both P1 and P2 are below 25 Pa·s.

## 5.6 Conclusion

Repeatability and reproducibility of slag viscosity measurements are demonstrated. Comparing slags with and without sulphur, the largest deviation of LV measurements obtained with different apparatuses is 0.11. A series of viscosity measurements and microstructural analyses of quenched samples indicate that blending coal ashes with petroleum coke ash has a complex effect on the viscosity. Adding petroleum coke ash to coal ashes changes the relative amounts of network formers and modifiers, as well as the liquidus temperature and types/quantity of solids in the resulting slags. Vanadium, a major component of petroleum coke ashes, has the particular effect of promoting the formation of spinel in slags below 1300 °C. These combined effects lead to an increased temperature range of operability for one coal ash tested, but do not significantly change the range of operability for the other coal ashes tested. Both petroleum coke ashes tested are suitable for operation at or above 1275 °C when they are not blended with coal ash. Phases predicted using FactSage did not match solid phases observed in quenched samples. Vanadium-rich slags ( $\geq 5$  wt.%  $V_2O_5$ ) require development of a better viscosity measurement technique due to high reactivity with spindle/crucible materials.

## 5.7 Acknowledgements

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## **Chapter 6. Interactions of vanadium-rich slags with crucible materials during viscosity measurements**

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## 6.1 Abstract

Slag chemistry is important for the assessment of flow behaviour of slags produced during gasification of coal and coal-petroleum coke blends. Slags containing vanadium species react readily with the crucible and spindle materials used for viscosity measurements. Interaction of vanadium-rich slags with various materials has been investigated in order to obtain a better understanding of the impact of containment materials on the resulting slag chemistry and viscosity.

The bulk and phase compositions of two petroleum coke slags in  $\text{Al}_2\text{O}_3$ , Mo, Pt and Ni crucibles produced under different laboratory conditions were analysed, and kinetics of slag composition changes at 1400 °C were determined. Mechanisms of the slag interactions with crucibles are described. They involve exchanging of crucible and slag constituents, formation of interfaces with distinct compositions, and continuously changing phase equilibria in the system. For slag processed in Ni and Pt crucibles, reduction of Fe and Ni from oxide to metallic form occurs and is followed by dissolution into the crucible materials.

Viscosity of slags with Mo, Ni and  $\text{Al}_2\text{O}_3$  crucibles are determined in the temperature range 1200-1500 °C. Resulting changes in the bulk composition of the processed slag has an impact on the slag viscosity. At given temperatures, viscosities of the slags produced in different crucibles are different. The impact of crucible materials and their applicability in viscosity measurements of high-vanadium containing slags are also discussed in order to define the optimal conditions.

*Keywords:* Slag, Viscosity, Petroleum coke, Gasification, Vanadium

## 6.2 Introduction

Petroleum coke (petcoke) ash can contain high concentrations of vanadium, nickel, and iron. Vanadium oxide is a major impurity in the petcoke slags which is not found in coal ash. Vanadium containing slags are typically produced during gasification of coal-petroleum coke blended feed. Petcoke ash chemistry depends on the petcoke's origin. Vanadium concentration in petcoke ash can be as high as 75 wt.% [1]. However, total ash content in petcoke is typically between 1-2 wt.%, while ash content in the coal is around 10 wt.%, and the total concentration of vanadium in the slags from coal-petcoke blends depends on the feedstock blending ratio. The oxidation state of vanadium depends upon the temperature and partial pressure of oxygen [2, 3]. Due to the low oxygen partial pressure which may be encountered in a gasifier ( $O_2$  activity =  $10^{-9}$ - $10^{-7}$  at temperatures  $T=1300$ - $1600$  °C [4]), a  $VO_x$  phase can precipitate during gasification. Previous studies have identified karelianite ( $V_2O_3$ ) in slag which may increase the viscosity of a coal/petcoke mixture [5]. Attempts to predict the effect of vanadium on viscosity have been made using the FactSage thermodynamic package [6, 7]. However, the FactSage databases do not consider vanadium oxide solutions which were observed in quenched samples [7]. A recent study of phase equilibria for synthetic coal-petcoke slag mixtures in simulated gasification conditions reported compositions of liquid and solid phases at 1500 °C [8]. This may provide some input in the modelling of slag viscosity; however the phase equilibria approach may not be suitable for predicting slag viscosity in vanadium-containing slags processed in industrial conditions as it will be explained later in the paper.

Despite the importance of real viscosity data for slags, experimentally the effect of vanadium on the viscosity of gasification slags is rarely studied [6, 7, 9]. Major obstacles associated with viscosity measurements of vanadium containing gasification slags are: i) transition of oxidation states of metals in precursor materials (ashes) during slag formation and ii) interaction of the slag with crucible materials, which may continuously

change phase equilibria in the system and slag bulk composition. Park and Oh [6] obtained viscosity results for slags from a Korean anthracite blended with CaO and V<sub>2</sub>O<sub>3</sub>, where V<sub>2</sub>O<sub>3</sub> content in the slags did not exceed 3.4 wt.%. They highlighted the effect of petcoke on the temperature of critical viscosity; the temperature below which viscosity sharply increases upon further cooling. The temperature of critical viscosity was higher in petcoke-coal ash blends compared with pure coal ashes. Viscosities of different petcoke-coal ash blends were measured in previous studies by the current authors [7, 9]. Slag viscosity data were obtained with a high level of confidence for compositions containing up to 5 wt.% V<sub>2</sub>O<sub>3</sub>. However, accurate viscosities of pure petcoke slags were difficult to obtain because the compositions of these slags were significantly affected by their interactions with crucible materials.

In the present study, interactions of vanadium-rich slags with metallic and refractory materials are investigated. This will provide a better understanding of the impact of metallic and refractory materials on slag chemistry. Conventional and potential crucible materials applicable to viscosity measurements of high-vanadium containing slags produced by gasification of petcoke and coal-petcoke blends are assessed. Results of this work also provide some insights into the reactions and phase transformations which can be associated to slag flow behaviour in industrial gasification conditions.

## **6.3 Experimental**

### **6.3.1 Sample preparation**

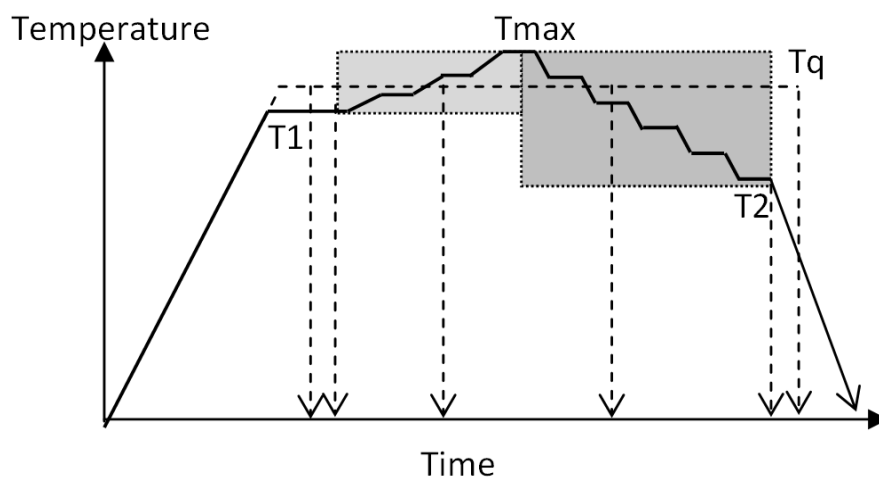
Two petcoke ashes and four crucible materials were used in this study. The petcoke ash compositions and crucible materials tested are given in Table 6.1. Crucible materials include metals: Mo (99+% purity), Ni (99%), Pt (99.99%), and high-density alumina ceramic (99.7% ). Pelletised petcoke ashes were melted in a reducing gas atmosphere (provided by a graphite enclosure) in flowing inert gas (nitrogen or argon). The partial pressure of oxygen,  $p_{O_2}$ , in the system has been measured using an oxygen probe (HT

oxygen probe, Australian Oxytrol Systems). At temperatures between 1200-1450 °C, O<sub>2</sub> activity values near the slag surface were detected between  $2.8 \times 10^{-11}$  –  $2.9 \times 10^{-9}$ .

**Table 6.1: Composition of petcoke ashes (wt.%) and crucibles used in the study**

<b>Composition</b>	<b>P1</b>	<b>P2</b>
SiO <sub>2</sub>	37.77	43.41
Al <sub>2</sub> O <sub>3</sub>	16.60	14.82
Fe <sub>2</sub> O <sub>3</sub>	7.35	9.41
TiO <sub>2</sub>	1.10	4.13
CaO	15.82	1.36
MgO	3.87	0.62
Na <sub>2</sub> O	1.67	1.13
K <sub>2</sub> O	0.96	1.75
V <sub>2</sub> O <sub>5</sub>	6.67	20.69
NiO	1.09	2.68
P <sub>2</sub> O <sub>5</sub>	<0.2	<0.2
SO <sub>3</sub>	7.11	<0.2
<b>Crucibles</b>	<b>P1</b>	<b>P2</b>
Al <sub>2</sub> O <sub>3</sub>	+	+
Mo	+	+
Ni	+	
Pt	+	

Some of the slag samples were obtained prior to and after viscosity measurements (described below). Their processing temperature profiles were similar to Profile 1 presented in Figure 6.1. Cold rod quenching experiments (Mo, Ni and Al<sub>2</sub>O<sub>3</sub> supports) or equilibration in envelopes (Pt support) were performed to obtain quenched slag samples at a fixed temperature (T<sub>q</sub>) for investigation of the kinetics of slag interaction with support materials. Cold rod quenching involves dipping a Mo or Al<sub>2</sub>O<sub>3</sub> rod in the molten slag, removing it and then cooling it in water. Envelope equilibration involves heating the slag sample in a small envelope made of the material of interest and suddenly dropping the envelope out of the furnace into water for quenching. Temperature profiles used for slag-crucible interaction kinetics are depicted by Profile 2 in Figure 6.1.



**Figure 6.1:** Temperature profiles used for slag preparation and processing. Profile 1 is in solid lines, Profile 2 and quenching are in dashed lines. The shaded area corresponds to viscosity measurement.

### 6.3.2 Sample analysis

Slowly cooled (post-viscosity tests) and quenched slags were analysed using Scanning Electron Microscopy (SEM) in backscattering mode. Energy Dispersive X-ray Spectroscopy (EDS) was undertaken to identify the elemental composition of selected regions of the slag. Electron Probe Microanalysis (EPMA) was used to identify the composition of solid and liquid phases in the slag samples. In reducing conditions, iron and vanadium are present in the slag as  $\text{Fe}^{2+}$  or  $\text{Fe}^{3+}$  and  $\text{V}^{2+}$  or  $\text{V}^{3+}$  respectively. EPMA analysis provides information on the cation ratio present in the sample. However the oxidation states of iron and vanadium in the various phases were not measured in this study. For ease of presentation in the present study all of the iron was recalculated to the ferric oxidation state and vanadium as  $\text{V}_2\text{O}_3$ .

Amounts of solids were determined using results of EPMA analysis (elements content in the phases) of the samples and mass balances for elements, which do not participate in slag-crucible interactions (such as Si and Ca). Bulk chemical composition of the slag and ash were determined by Inductively Coupled Plasma - Atomic Emission Spectroscopy (ICP-AES) or X-ray fluorescence (XRF), and confirmed by EPMA.

### **6.3.3 Viscosity measurements**

Slag viscosity measurements were performed in a Carbolite 3 litre BLF1700 furnace with a Brookfield RVDV-IIIU rheometer [7]. Petcoke ashes were placed into the crucible and heated in a flow of argon to form liquid slag. To ensure reducing conditions, the crucible was placed in a sacrificial graphite liner.  $O_2$  activity values are specified in section 6.3.1. Viscosity measurements were performed using a rotating bob (made from Mo or  $Al_2O_3$ ) connected to the rheometer. Measurements were conducted at incremental temperature steps (20-30 °C) within the temperature range 1200-1500 °C. Temperature profiles where viscosity measurements were taken are marked in Figure 1 (shaded areas). After slag pre-melting at  $T_1 = 1225-1300$  °C (sample heating rate is 5°C/min), the temperature of the samples was increased to  $T_{max} = 1340-1500$  °C. During the cooling stage (2°C/min), samples were held at each temperature ( $T_2$ ) for at least 20-30 min intervals to equilibrate temperature inside the crucible. At each temperature, the slag viscosity was measured at least three times with various shear rates to ensure the repeatability of measurements. The standard deviation of the measurements was below 5% of the mean value.

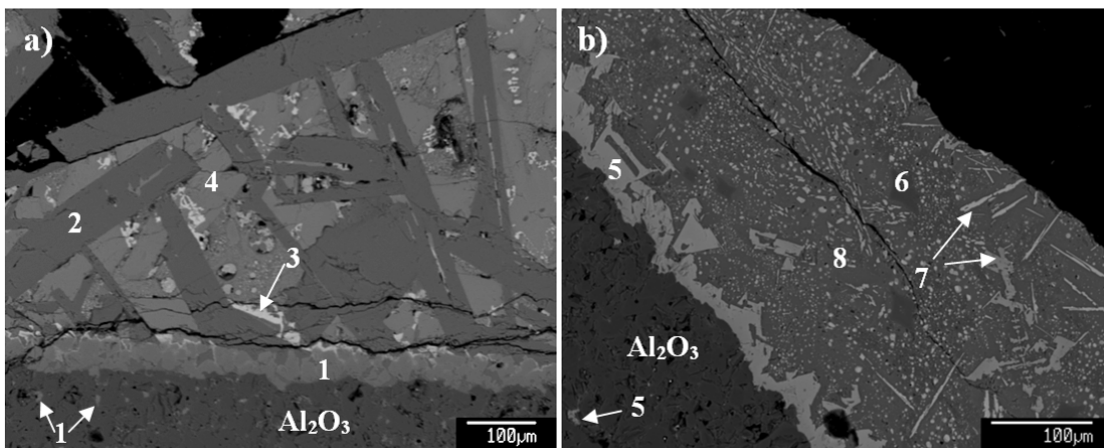
## **6.4 Results and discussion**

Both petcoke slags were highly reactive with the crucibles used in this study. Interactions of the slags with different materials were investigated in terms of the phase formations, kinetics and mechanisms of reaction. For both petcoke, results with each crucible material are presented below.

### **6.4.1 Slag processed in $Al_2O_3$**

Figure 6.2 shows micrographs of the petcoke slags in an  $Al_2O_3$  crucible following Profile 1 heat treatment with slow cooling. Phase compositions observed in Figure 6.2 are listed in Table 6.2. Both samples have a clear interface phase which was identified as  $Al_2O_3$  spinel. The concentration of V and Ni are higher in the interface observed for P2 than for

P1. This is expected since P2's initial composition has higher concentrations of both Ni and V. V-rich spinels are also observed with both slags, however, only Ni concentration is higher in slag P2's V-spinel than in slag P1's V-spinel. Differences in the initial petcoke ash Ca content result in differences for the other observed phases: feldspar was formed in slag P1, while some mullite appeared in P2 slag. Molten slag has penetrated into the  $\text{Al}_2\text{O}_3$  support (up to 2 mm in depth) and formed  $\text{Al}_2\text{O}_3$  spinel in the pores between  $\text{Al}_2\text{O}_3$  grains. Composition of these spinels determined by EPMA is very similar to the composition of the interface phases ((1) and (5) in Figure 6.2) observed with each petcoke slag. The observed microstructures in slow-cooled slags resulted from compositional changes and reactions which occur during the whole heat treatment in temperature Profile 1 and do not indicate at which stage of the treatment the phase changes occurred.



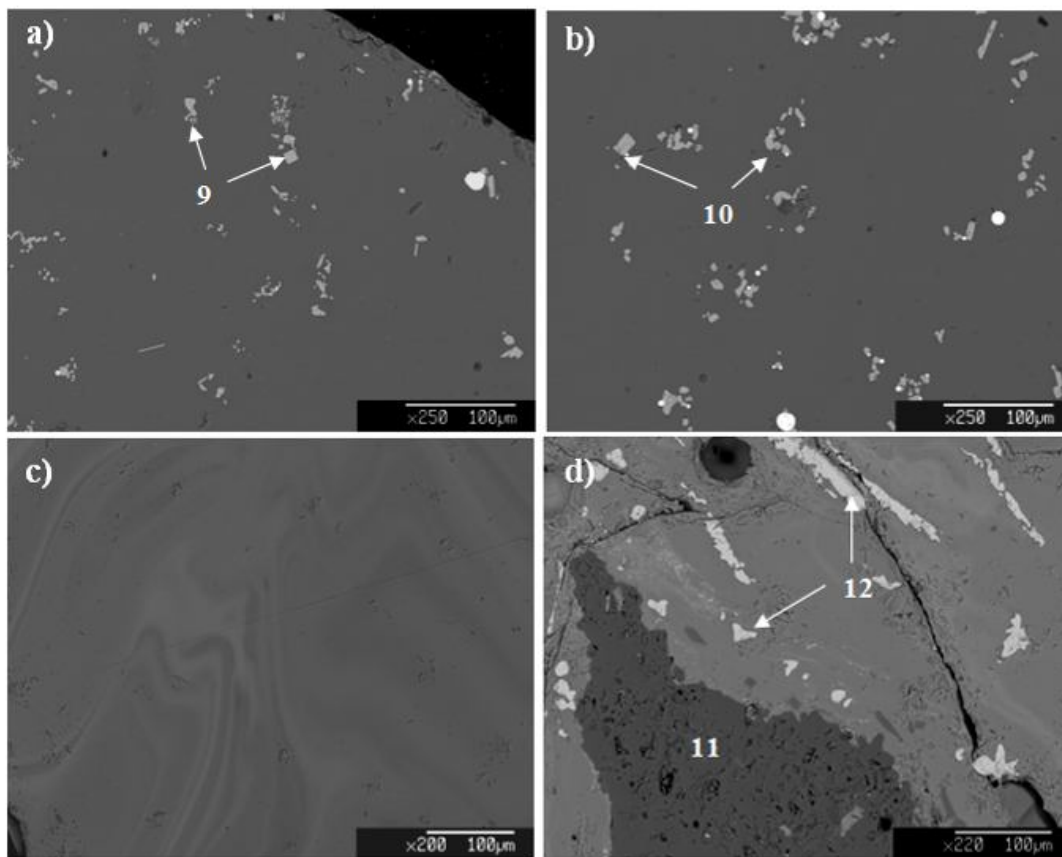
**Figure 6.2:** Microstructures of slow cooled P1 (a) and P2 (b) slags with  $\text{Al}_2\text{O}_3$  supports after processing with Profile 1. Numbered phases are listed in Table 6.2.

**Table 6.2: Phase compositions (wt.%) of P1 and P2 slags with Al<sub>2</sub>O<sub>3</sub> crucibles**

	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	CaO	MgO	TiO <sub>2</sub>	K <sub>2</sub> O	Na <sub>2</sub> O	V <sub>2</sub> O <sub>3</sub>	Cr <sub>2</sub> O <sub>3</sub>	NiO
<b>P1-alumina: slow cooled</b>											
Al-spinel (1)	<0.1	63.09	10.34	<0.1	15.43	<0.1	<0.1	<0.1	1.14	0.21	10.77
Feldspar (2)	45.24	33.71	0.49	17.74	0.11	<0.1	0.19	1.03	<0.1	<0.1	<0.1
V-spinel (3)	2.14	7.32	51.21	0.94	6.26	1.89	0.49	0.74	18.31	3.10	7.54
Former liquid (4)	37.78	10.75	8.72	22.10	8.98	2.12	<0.1	0.41	5.66	0.98	1.54
<b>P2-alumina: slow cooled</b>											
Al-spinel (5)	<0.1	53.61	16.92	<0.1	3.05	<0.1	<0.1	<0.1	3.03	0.57	25.12
Mullite (6)	25.05	65.25	3.36	<0.1	0.15	0.38	<0.1	<0.1	4.35	0.75	0.52
V-spinel (7)	0.40	46.29	26.74	<0.1	4.38	0.15	<0.1	<0.1	6.95	1.20	16.49
Former liquid (8)	61.16	14.61	2.82	3.65	0.62	0.44	2.82	1.19	9.74	1.59	0.35
<b>P1-alumina: quenched from 1400°C (1 hour)</b>											
Former liquid	44.26	20.87	2.78	19.11	4.44	1.17	1.17	2.12	2.84	0.2	<0.1
V-rich (9)	<0.1	2.81	1.66	0.70	1.12	3.87	<0.1	<0.1	89.65	<0.1	<0.1
<b>P1-alumina: quenched from 1400°C (12 hours)</b>											
Former liquid	42.69	25.35	2.72	18.17	3.68	1.10	1.07	2.06	2.09	0.3	<0.1
V-rich (10)	1.35	4.79	1.32	0.99	0.82	5.20	<0.1	<0.1	85.35	<0.1	<0.1
<b>P2-alumina: quenched from 1400°C (1 hour)</b>											
Former liquid	47.19	18.06	9.58	4.2	1.15	0.72	1.40	1.29	14.01	2.47	2.34
<b>P2-alumina: quenched from 1400°C (12 hours)</b>											
Former liquid	49.48	21.62	7.15	4.08	1.08	0.59	1.72	1.78	10.48	1.95	1.90
Mullite (11)	29.83	51.77	1.39	0.34	<0.1	<0.1	0.10	<0.1	0.55	<0.1	0.14
V-rich (12)	<0.1	7.24	9.34	<0.1	0.34	1.15	<0.1	<0.1	63.91	11.50	6.37

In order to investigate the kinetics of the slag-Al<sub>2</sub>O<sub>3</sub> reaction at a given temperature, slag samples were quenched from a fixed temperature after being held at that temperature for 1 to 24 h. A quenching temperature of 1400 °C was taken as an intermediate temperature used in viscosity measurements. As shown in Figure 6.3a and 6.3b, quenched P1 slag always has some amount of a solid phase which was identified as V-rich oxide. The amount of solid phase was nearly the same (between 3.5-5%) for all quenched P1 slags. The P2 sample quenched after 1 h of equilibration only has a non-homogeneous liquid phase (Figure 6.3c). Mullite was identified in the samples quenched after 3 h of equilibration, while both mullite and V-rich spinel were observed in the samples quenched after 6 and 12 h (Figure 6.3d) of equilibration. Phase compositions for the P1 and P2 samples quenched after 1 and 12 h of equilibration are reported in Table 6.2.

Compositional changes in the quenched samples are mainly associated with changes in V and Al content in the solid and liquid phases, as illustrated in Figure 6.4. Major changes in concentrations of V and Al oxides occur in the first 3 h of equilibration. However, continuous changes of  $\text{Al}_2\text{O}_3$  and V content in the slag compositions demonstrate that phase equilibria has not been achieved even after 12 h of equilibration. It is clear that the  $\text{Al}_2\text{O}_3$  crucible supplies extra Al in the melt while V is incorporated into the crucible. With longer equilibration time, the slopes become flat which indicates a lower rate of V and Al exchange between the slags and crucibles.



**Figure 6.3: Microstructures of P1 and P2 slags on  $\text{Al}_2\text{O}_3$  supports quenched from 1400 °C: P1 after 1h (a), P1 after 12 h (b), P2 after 1h (c), and P2 after 12h (d).**

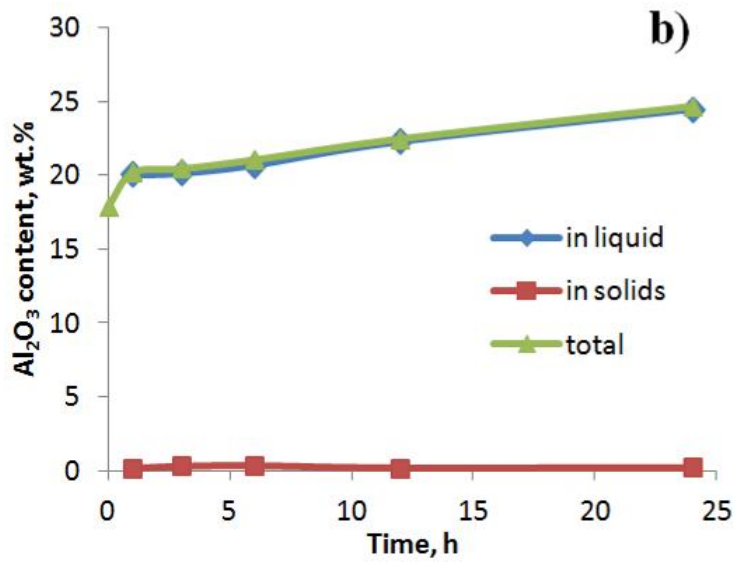
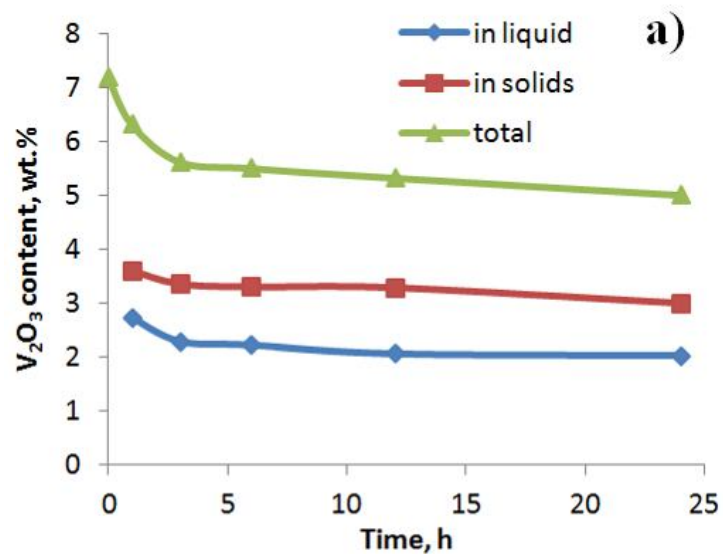


Figure 6.4:  $V_2O_3$  and  $Al_2O_3$  content in P1 (a, b) and in P2 (c, d) slags on  $Al_2O_3$  supports quenched from 1400 °C. (part 1 of 2)

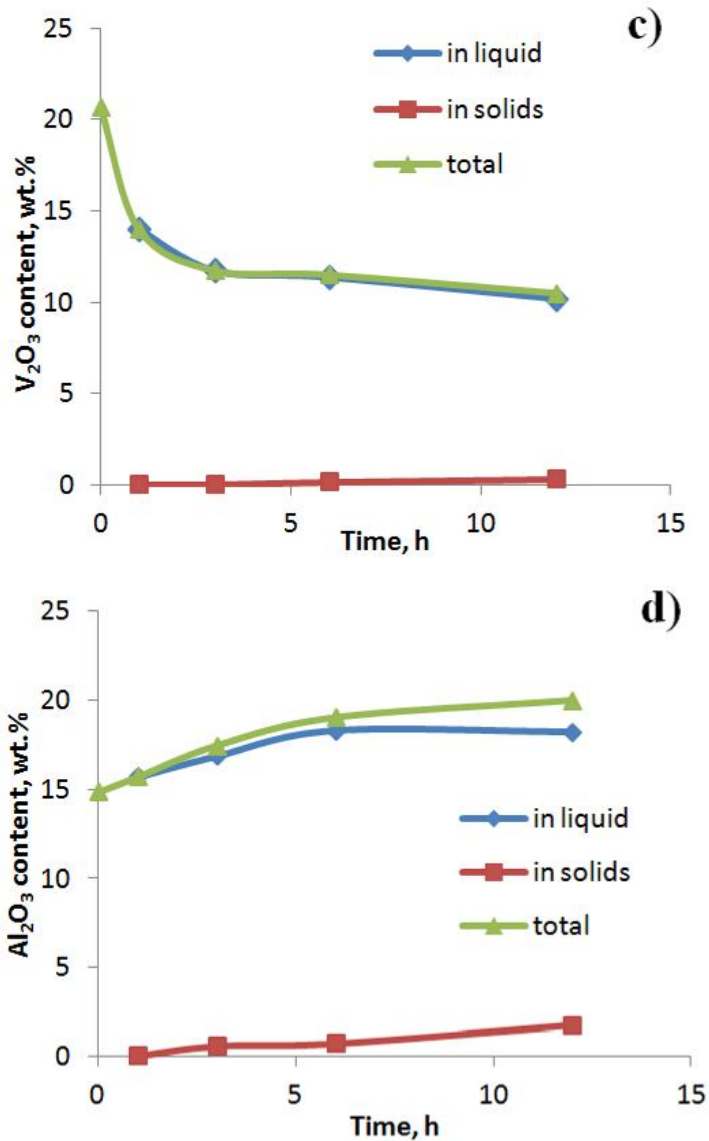
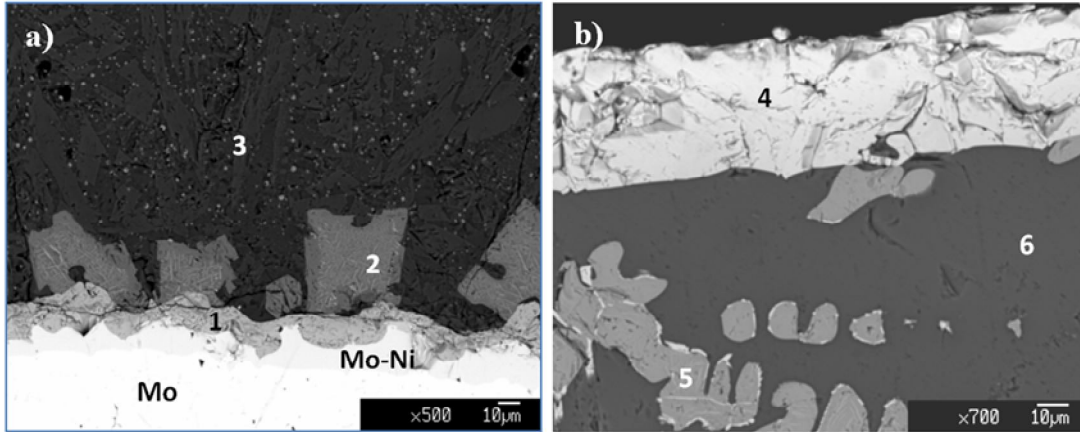


Figure 6.4: V<sub>2</sub>O<sub>5</sub> and Al<sub>2</sub>O<sub>3</sub> content in P1 (a, b) and in P2 (c, d) slags on Al<sub>2</sub>O<sub>3</sub> supports quenched from 1400 °C. (part 2 of 2)

#### 6.4.2 Slag processed in Mo

Micrographs of P1 and P2 slags processed in Mo crucibles with Profile 1 are shown in Figure 6.5. The identified oxide phases are listed in Table 6.3. With P1, the corroded Mo crucible dissolved metallic Ni from the slag and formed a layer 10-30 μm thick on the edge of the Mo crucible (Figure 6.5a). For both P1 and P2, there is a clear interface phase formed between the crucible (not attached to the P2 sample) and the slag, which is identified as a Mo and V oxides solid solution, which contains small concentrations of

NiO, Cr<sub>2</sub>O<sub>3</sub> and TiO<sub>2</sub>. Other solid phases observed are V-rich spinel (grey 10-50 μm crystallites in Figure 6.5a and 6.5b) and dispersed inclusions of Mo oxide (light grey particles in Figure 6.5a). As P2 has a higher V content (approximately three times higher) than P1, the concentration of V at the interface is also higher (two times). More Mo was precipitated in the V-rich oxide of sample P1, which also formed closer to the Mo-rich interface.



**Figure 6.5:** Microstructures of slow cooled slags P1 (a) and P2 (b) on Mo supports after processing with Profile 1. Numbered phases are listed in Table 6.3.

**Table 6.3: Phase compositions (wt.%) of P1 and P2 slags with Mo crucibles**

	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	CaO	MgO	TiO <sub>2</sub>	K <sub>2</sub> O	Na <sub>2</sub> O	V <sub>2</sub> O <sub>3</sub>	Cr <sub>2</sub> O <sub>3</sub>	NiO	MoO <sub>3</sub>
<b>P1-Mo: slow cooled</b>												
Interface (1)	<0.1	<0.1	0.27	<0.1	<0.1	0.93	<0.1	<0.1	10.89	1.00	0.54	78.95
V- rich (2)	<0.1	5.21	18.43	<0.1	3.78	1.46	<0.1	<0.1	49.04	3.23	2.53	20.66
Former liquid (3)	44.40	19.24	5.95	15.64	3.12	1.08	1.07	2.39	2.98	0.35	0.25	3.30
<b>P2-Mo: slow cooled</b>												
Interface (4)	<0.1	<0.1	<0.1	<0.1	<0.1	0.64	<0.1	<0.1	21.73	1.94	0.21	79.59
V-rich (5)	<0.1	3.25	14.20	<0.1	0.39	1.35	<0.1	<0.1	58.94	4.42	6.82	5.98
Former liquid (6)	64.38	16.31	3.74	3.45	0.87	0.25	3.04	1.43	1.36	0.23	0.42	2.72
<b>P1-Mo: quenched from 1400°C (1 hour)</b>												
V-rich (7)	0.5	3.66	2.08	0.4	0.89	3.19	<0.1	<0.1	77.91	6.23	<0.1	0.23
Former liquid	47.23	18.99	3.78	17.48	4.09	1.10	1.46	2.37	2.58	0.33	<0.1	<0.1
<b>P1-Mo: quenched from 1400°C (12 hours)</b>												
V-rich (8)	<0.1	2.51	2.66	<0.1	0.84	3.78	0.03	0.00	80.45	6.05	0.10	3.05
Former liquid	44.64	18.84	5.44	18.42	4.50	1.20	1.18	2.00	2.62	0.30	<0.1	0.21
<b>P2-Mo: quenched from 1400°C (1 hour)</b>												
V-rich (9)	<0.1	3.93	10.24	<0.1	0.35	0.93	<0.1	<0.1	64.90	11.72	6.64	1.11
Former liquid	45.30	16.00	7.56	4.14	1.21	0.5	1.56	1.34	13.97	2.47	2.18	3.59
<b>P2-Mo: quenched from 1400°C (12 hours)</b>												
V-rich (10)	<0.1	3.63	10.77	<0.1	0.43	1.11	<0.1	<0.1	65.44	11.82	5.14	1.54
Former liquid	49.16	16.32	7.34	4.56	1.35	0.53	1.65	1.49	8.28	1.47	1.47	6.03

As was done for the Al<sub>2</sub>O<sub>3</sub> tests, quenching experiments were conducted to investigate the kinetics of slag-Mo crucible interaction. Samples were quenched after 15 min, 1 h, 3 h, 6 h, 12 h and 24 h of equilibration at 1400 °C. All P1 and P2 quenched slags consist of a liquid phase and a V-rich spinel. Phase compositions of samples quenched after 1 and 12 h of equilibration are listed in Table 6.3 and shown in Figure 6.6. The quantity of solids in the P1 slag does not change dramatically over 24 h, while gradual growth of solids occurs with greater equilibration time for the P2 slag (Figure 6.7). The V and Mo content in all solids, in the liquid phase, and in total for quenched P1 and P2 slags processed in a Mo crucible are shown in Figure 6.8. In both P1 and P2 slags, the initial concentration of V decreases rapidly in the first hour of equilibration. The V content is partially restored after 6 and 24 h of equilibration for P1 and P2 slags, respectively. After 6 h of equilibration, the concentration of V in P1 slag continues to decrease. Over this time, the Mo concentration in P1 slag increases from 0.4 to 3.3 wt.%. For P2 slag, the concentration of Mo rapidly increases in the first hour of equilibration, reaches a

maximum of 6.3 wt.% after 6 h, and then slightly decreases in the next 18 h. The difference in the kinetics of V depletion and Mo enrichment in P1 and P2 slags is likely associated with the concentration of V in the initial slag. At first, the P2 slag reacts more aggressively with the Mo support. This is seen by the rapid increase in liquid Mo content. Past 6 h equilibration time, the Mo-V-oxide interface forms a continuous layer, as was observed with the slowly cooled slag (Figure 6.5b). This phase still consumes Mo from the support but resists penetration of Mo in the liquid slag. Moreover this phase may also consume already dissolved Mo from the liquid slag (Mo decrease in bulk slag) and V which was dissolved in the Mo support earlier. The Mo-V-oxide interface is also forming in P1 slag but V dissolution into the crucible and Mo uptake into the slag are slower for the first 12 h. Past 12 h, the V near the interface likely becomes depleted and Mo from the support is no longer consumed by the interface phase growth. The discontinuous interface phase (as observed in Figure 6.5a) allows the Mo to penetrate into the bulk slag.

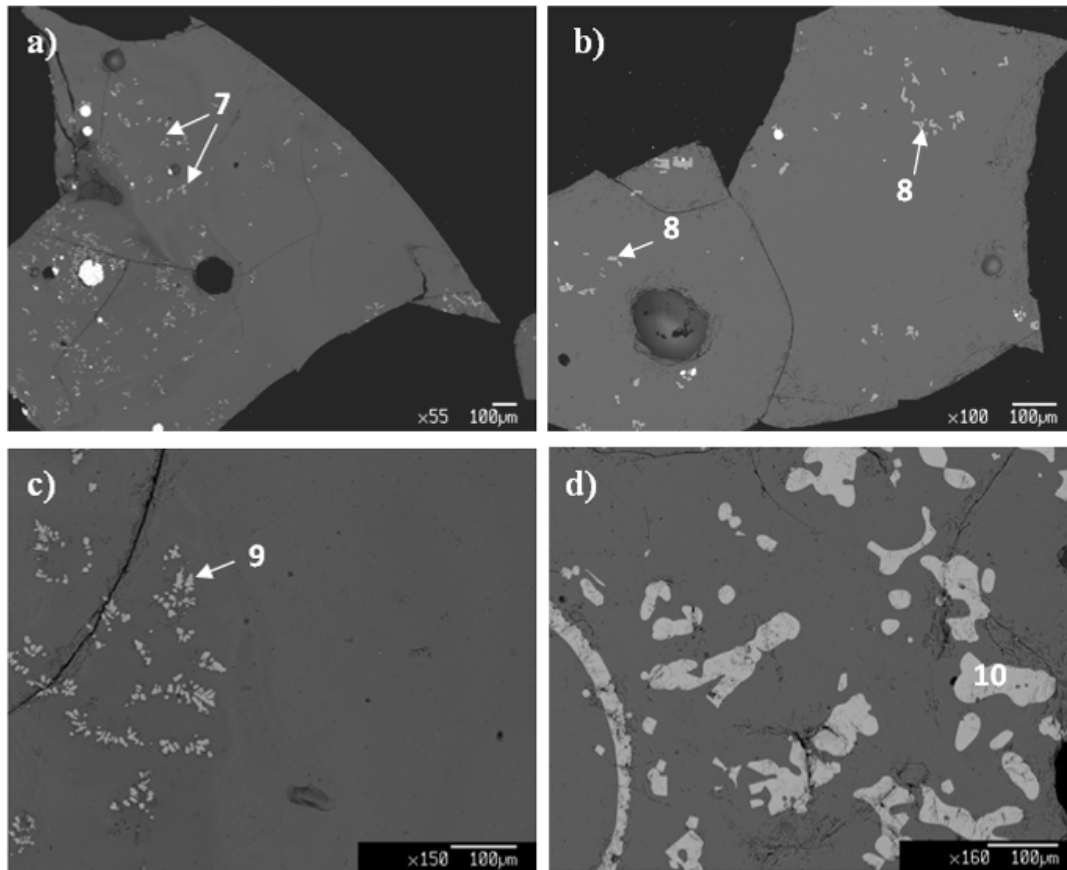


Figure 6.6: Microstructures of P1 and P2 slags on Mo supports quenched from 1400 °C: P1 after 1h (a), P1 after 12 h (b), P2 after 1h (c), and P2 after 12h (d).

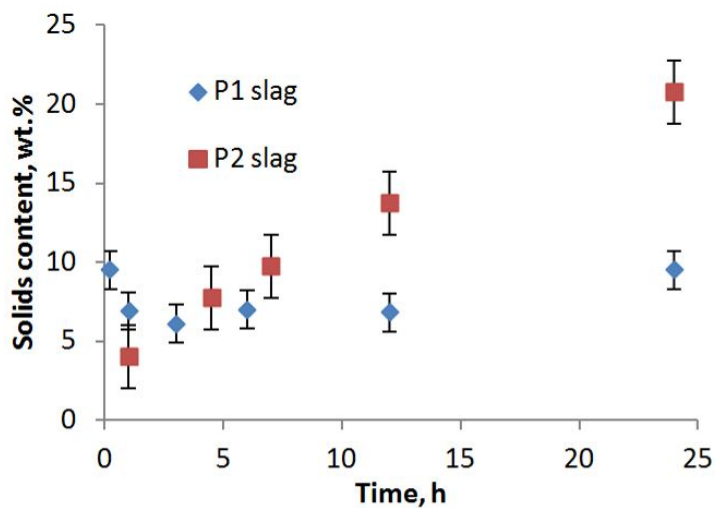


Figure 6.7: Solid phase content in P1 and P2 slags on Mo supports quenched from 1400 °C.

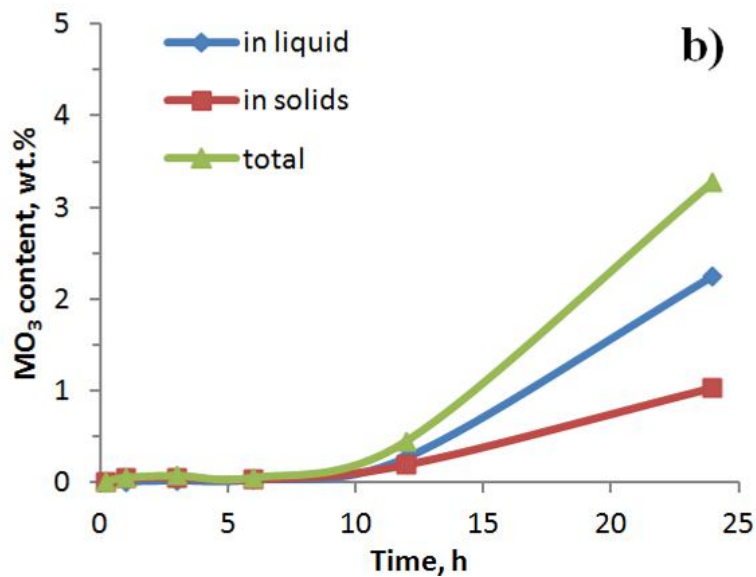
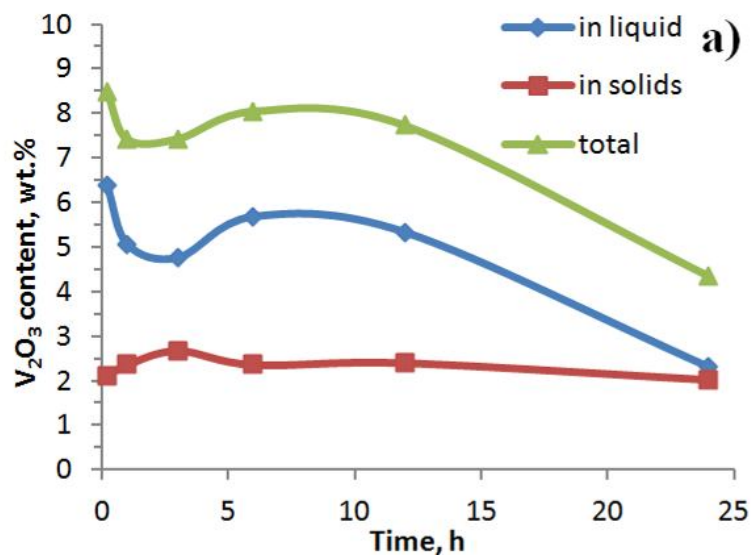


Figure 6.8:  $V_2O_3$  and  $MoO_3$  content in P1 (a, b) and in P2 (c, d) slags on Mo supports quenched from 1400 °C. (part 1 of 2)

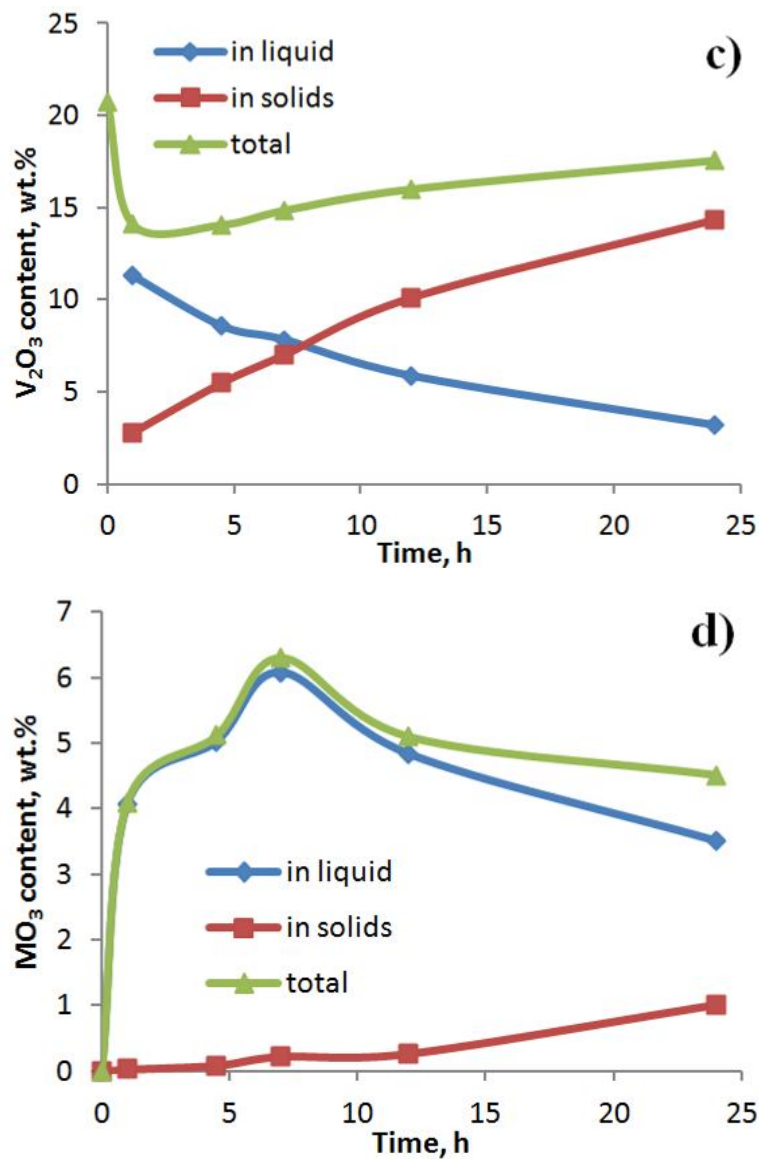


Figure 6.8: V<sub>2</sub>O<sub>3</sub> and MoO<sub>3</sub> content in P1 (a, b) and in P2 (c, d) slags on Mo supports quenched from 1400 °C. (part 2 of 2)

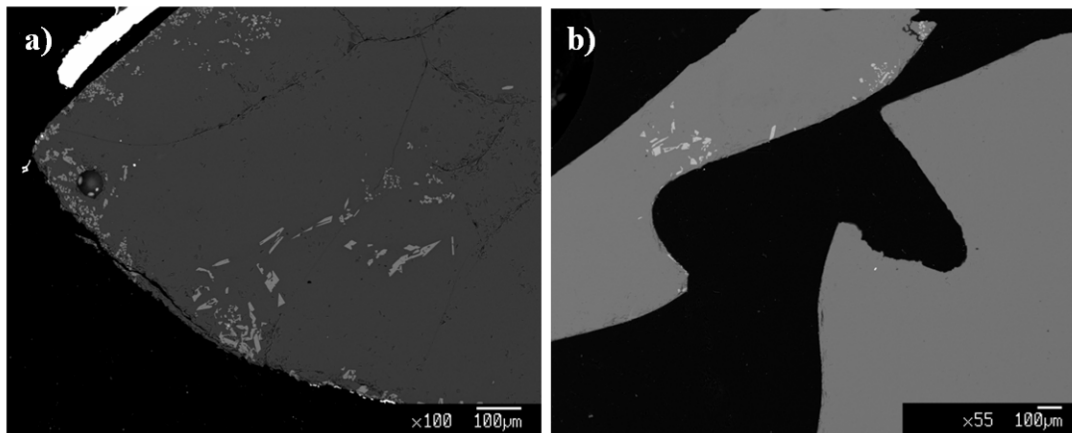
### 6.4.3 Slag processed in Pt or Ni

Pt and Ni were considered as alternative candidates for crucibles to be used in V-rich slags preparation and viscosity measurements. However, both metals were found to react with V-rich slag.

Quenched P1 slags from Pt crucibles were obtained after equilibration at 1400 °C ( $O_2$  activity =  $2 \times 10^{-9}$ ) for 1 and 12 h. Their compositions are listed in Table 6.4 and the microstructures are shown in Figure 6.9. Most of the changes in composition occurred in the first hour of equilibration. The slag lost significant amounts of Fe and Ni, as well as some V. EDS of the Pt crucibles confirmed that these metals were dissolved in the crucibles. With longer equilibration time, V content in the slag was further reduced. This led to a reduction of solids (V-based oxides) in the slag which are nearly all gone after 12 h.

**Table 6.4: Phase compositions (wt.%) of P1 slags with Pt crucibles**

	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	CaO	MgO	TiO <sub>2</sub>	K <sub>2</sub> O	Na <sub>2</sub> O	V <sub>2</sub> O <sub>3</sub>	Cr <sub>2</sub> O <sub>3</sub>	NiO
<b>P1-Pt: quenched from 1400°C (1 hour)</b>											
V-rich	<0.1	3.08	0.64	<0.1	1.85	2.52	<0.1	<0.1	84.86	6.39	<0.1
Former liquid	43.4	19.89	0.36	21.73	5.14	1.26	1.00	2.12	4.33	0.5	<0.1
<b>P1-Pt: quenched from 1400°C (12 hours)</b>											
V-rich	<0.1	3.38	0.27	<0.1	1.54	3.24	<0.1	<0.1	85.32	5.8	<0.1
Former liquid	43.9	20.75	0.25	22.1	5.27	1.34	0.99	1.84	3.08	0.39	<0.1



**Figure 6.9: Microstructures of P1 slags equilibrated with Pt supports at 1400 °C for 1 h (a) and 12 h (b).**

Quenched P1 slags from Ni crucibles were obtained after 0.5 and 1.5 h at 1320 °C prior to viscosity measurements ( $O_2$  activity =  $4 \times 10^{-9}$ ). Two additional P1 slag samples were obtained after completion of the viscosity measurement. One sample was obtained by quenching slag (cold rod quenching) from 1390 °C and the second one was relatively

slow quenched (100 °C/min) from 1300 °C. Microstructures and compositions of these slags are presented in Figure 6.10 and Table 6.5, respectively. Composition of the liquid slag was affected by the reducing conditions of the system: Fe and Ni contents in the liquid slag significantly decreased with time by reduction of these oxides to their metallic form (indicated in Figure 6.10a). Liquid slag after the viscosity measurements (Figure 6.10c) has no Ni and a very small amount of Fe in oxide form. Some of the Fe and Ni are dispersed in the slag in metallic form. However, most of the Fe and Ni and some V were dissolved in the Ni crucible (determined by EDS of the crucible). It was observed that during heating to  $T_{\max}$  (1400 °C), the Ni support melted. This must be due to the incorporation of Fe and V (and some Mo from the viscometer's rotating bob) which reduced the melting point of the crucible (the melting point of pure Ni is 1455 °C). The resulting microstructures (Figure 6.10d) include the Ni (Fe, Mo, V) crucible, a former liquid slag phase, feldspar, V-rich oxide and an interface phase between the slag and crucible. The interface phase is identified as V-rich oxide containing Cr and Mo.

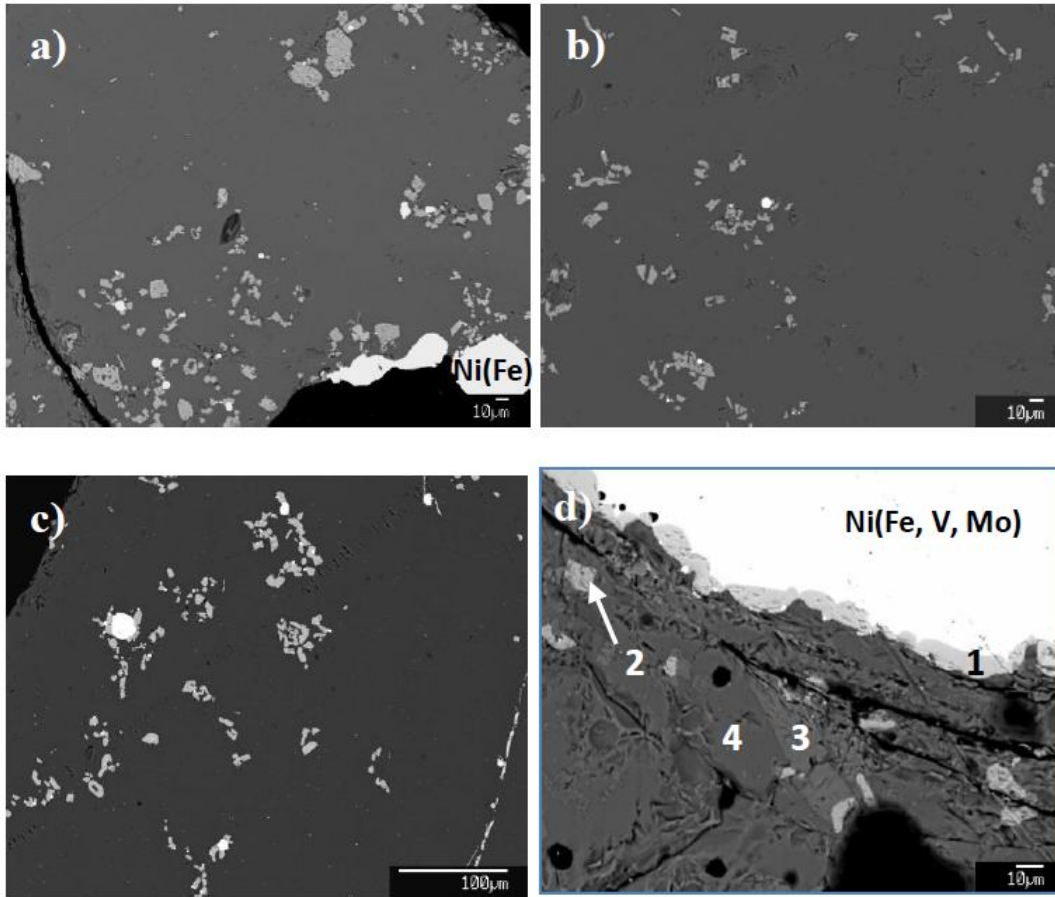


Figure 6.10: Microstructures of P1 slags processed with Ni. Slag was quenched after 0.5 h (a) and 1.5 h (b) at 1320 °C. Slag was also quenched from 1390 °C during Profile 1 (c) and slowly cooled after Profile 1 (d). Numbered phases are listed in Table 6.5.

Table 6.5: Phase compositions (wt.%) of P1 slags with Ni crucibles

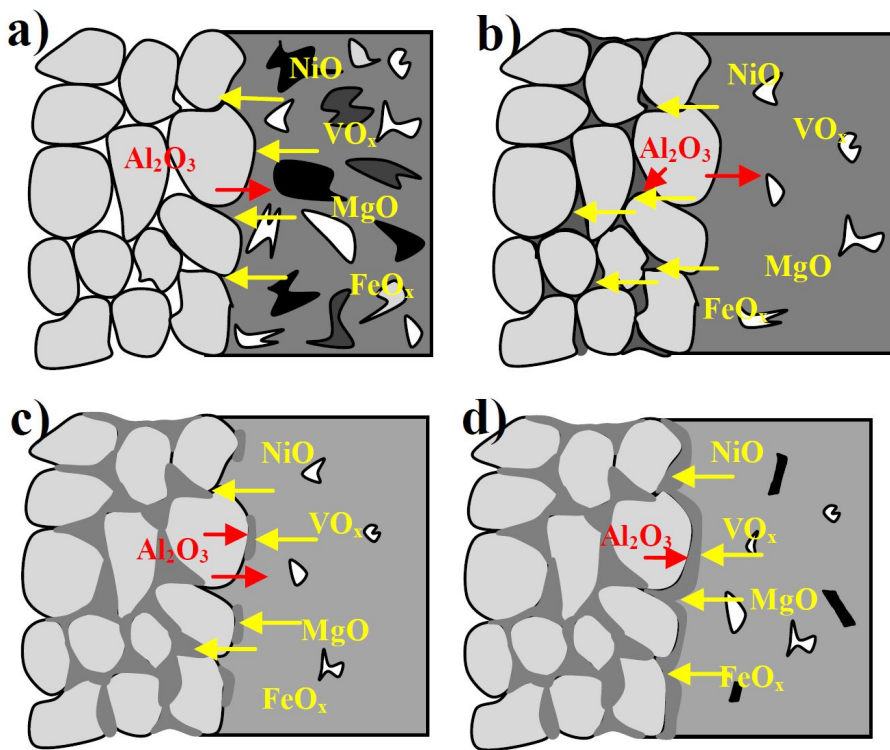
	SiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	CaO	MgO	TiO <sub>2</sub>	K <sub>2</sub> O	Na <sub>2</sub> O	V <sub>2</sub> O <sub>3</sub>	Cr <sub>2</sub> O <sub>3</sub>	NiO	MoO <sub>3</sub>
<b>P1-Ni: quenched from 1320°C (0.5 hour)</b>												
V-rich	<0.1	10.44	17.04	<0.1	9.21	0.81	<0.1	<0.1	50.87	9.29	1.83	<0.1
Former liquid	43.85	19.42	4.50	18.65	4.07	1.28	1.20	2.22	3.56	0.64	0.17	<0.1
<b>P1-Ni: quenched from 1320°C (1.5 hours)</b>												
V-rich	<0.1	2.28	1.60	<0.1	0.63	3.67	<0.1	<0.1	76.60	14.09	0.22	<0.1
Former liquid	46.00	20.54	1.75	19.66	4.61	1.11	1.28	2.22	2.19	0.42	<0.1	<0.1
<b>P1-Ni: quenched from 1390°C (post viscosity run)</b>												
V-rich	<0.1	2.27	<0.1	<0.1	<0.1	6.74	<0.1	<0.1	76.06	13.77	0.10	0.17
Former liquid	46.94	21.14	0.29	19.92	4.76	1.03	1.32	2.34	1.83	0.33	<0.1	<0.1
<b>P1-Ni: slow quenched from 1300°C (post viscosity run)</b>												
Interface (1)	<0.1	<0.1	0.46	<0.1	<0.1	0.28	<0.1	<0.1	81.12	14.47	0.31	2.88
V-rich (2)	<0.1	2.18	0.32	<0.1	0.61	6.28	<0.1	<0.1	76.01	13.82	<0.1	<0.1
Feldspar (3)	45.06	33.85	<0.1	18.73	0.36	<0.1	0.22	1.36	0.21	<0.1	<0.1	<0.1
Former liquid (4)	47.84	19.44	0.77	19.81	5.19	1.10	1.67	2.75	1.08	0.21	<0.1	<0.1

Fe, Ni and V are main constituents of the V-rich phase which appeared in slags from  $\text{Al}_2\text{O}_3$  and Mo crucibles (listed in Tables 6.2 and 6.3). Extraction of Fe and Ni, and partially of V, from the slags in Pt and Ni crucibles reduces the stability of the V-rich phase. This leads to changes in phase equilibria in the systems and likely decreases the liquidus temperature of the slags. However, P1 slag was less affected by these changes when processed with a Ni crucible rather than a Pt crucible.

#### **6.4.4 Mechanisms of slag-support interactions**

Based on the observation of slag microstructures and analysis of the compositional changes in the petcoke slags, proposed mechanisms of the slag-crucible interactions are described.

For the slags in  $\text{Al}_2\text{O}_3$  crucibles, the proposed mechanisms of interaction are illustrated in Figure 6.11. First, V-rich melt starts to penetrate between  $\text{Al}_2\text{O}_3$  grains (Figure 6.11a). The reaction of the penetrated melt leads to dissolution of  $\text{Al}_2\text{O}_3$  into the slag as well as formation of V-spinels in the pockets (1-10  $\mu\text{m}$ , as shown in Figure 6.2) between alumina grains (Figure 6.11b). The slag still may contain some V-rich solid phases (depending on the temperature) while a V-spinel interface is forming between the  $\text{Al}_2\text{O}_3$  crucible and slag (Figure 6.11c). With longer equilibration time and/or during the slag cooling, the spinel forms a continuous interface layer (Figure 6.11d). This interface spinel minimises V penetration into the  $\text{Al}_2\text{O}_3$  crucible. However,  $\text{Al}_2\text{O}_3$  enrichment in the bulk slag was observed over all 24 h tested. This led to changes in the amount and composition of V-rich solids, the composition of the liquid phase and appearance/or disappearance of other solid phases (such as mullite).



**Figure 6.11: Mechanisms of slag interaction with  $\text{Al}_2\text{O}_3$  supports. Red and yellow arrows indicate migration of support and slag constituents respectively and the targeting areas. Different contrast of liquid slag illustrates the compositional changes.**

For slags on Mo supports, the mechanisms of slag-crucible interaction are different. With slag melting, Ni and likely V dissolve in the Mo crucible (Figure 6.12a) and form a metallic interface (Figure 6.12b), while Mo is oxidised and dissolves into the slag. As melting progresses, the slag still contains some V-rich solids. Mo-V-oxide forms an interface on the surface Mo containing residual Ni and V. (Figure 6.12c). This interface finally forms a continuous layer, consuming Mo from the crucible and V from the metallic slag-crucible interface and from the molten slag (Figure 6.12d). The solid phase content in the slag (mainly V-rich oxide) depends on the initial slag bulk composition and the temperature. Since the bulk composition continuously changes over 24 h of equilibration, the solid phase content and solid phase composition change accordingly.

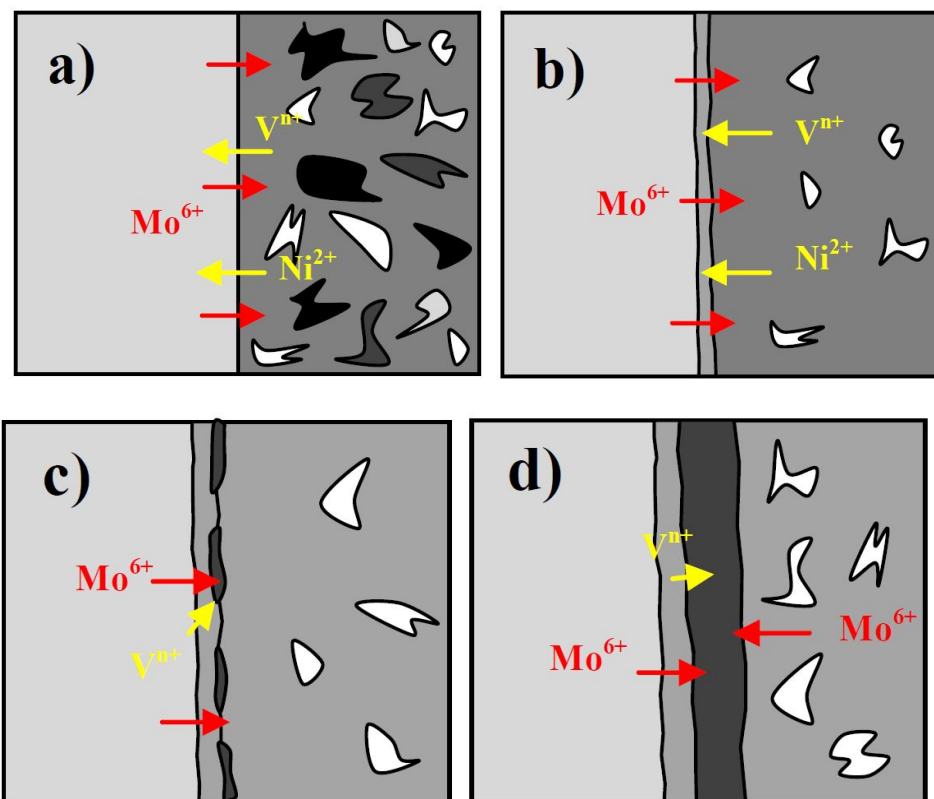


Figure 6.12: Mechanisms of slag interaction with Mo supports. Red and yellow arrows indicate migration of support and slag constituents respectively and the targeting areas.

For slags with Pt and Ni supports, the mechanisms of interaction include the reduction of Fe and Ni oxides in the slag to their metallic form. Much of the Fe and Ni, as well as some V, was taken up by the crucibles. Most of the Fe and Ni was lost within the first 1-1.5 h. V loss was slower. This is likely due to the fact that it is mainly present as a solid in the conditions tested.

#### 6.4.5 Bulk compositional changes and viscosity

As described above, the composition of the petcoke slags continuously changed during the viscosity measurements. Table 6.6 shows the bulk slag compositions prior to and after viscosity measurements. Compositions of slags processed in  $\text{Al}_2\text{O}_3$  for viscosity tests present a deficit of  $\text{V}_2\text{O}_3$ , which decrease to 3.9 wt.% in P1 slag and to 10.2 wt.% in P2 slag. Both slags accumulate  $\text{Al}_2\text{O}_3$  from the crucibles. While  $\text{Al}_2\text{O}_3$  content in P2 slag

increases significantly from 14.8 wt.% to 20.0 wt.% (42% increase from the initial content), Al<sub>2</sub>O<sub>3</sub> content in P1 slag increase by 2.5 wt.% only (14% of the initial content).

V<sub>2</sub>O<sub>5</sub> content after viscosity measurements for slags in Mo crucibles decreases by 3 wt.% and 7 wt.% (or 50% and 35% of the initial vanadium content) for P1 and P2, respectively. In these slags, Mo content reached 5.2 wt.% in P1 slag and 10.6 wt.% in P2 slag.

**Table 6.6: Composition of V-rich slags before (P1 and P2) and after viscosity tests. Key differences are in different fonts. Bold font shows enrichment, underlined are loses**

	<b>P1</b>	<b>P1-Mo</b>	<b>P1- Al<sub>2</sub>O<sub>3</sub></b>	<b>P1-Ni</b>	<b>P2</b>	<b>P2-Mo</b>	<b>P2- Al<sub>2</sub>O<sub>3</sub></b>
SiO <sub>2</sub>	40.7	41.3	39.6	44.5	43.4	42.3	47.5
Al <sub>2</sub> O <sub>3</sub>	17.9	17.9	<b>20.4</b>	19.7	14.8	13.8	<b>20.1</b>
Fe <sub>2</sub> O <sub>3</sub>	7.91	7.53	7.9	<u>1.49</u>	9.41	9.56	7.23
CaO	17.0	16.0	16.8	18.7	4.13	3.72	3.52
MgO	4.87	3.65	3.48	4.29	1.36	1.13	1.73
TiO <sub>2</sub>	1.19	1.03	1.15	1.31	0.62	0.52	0.32
Na <sub>2</sub> O	1.8	1.81	1.78	1.95	1.13	1.12	1.06
K <sub>2</sub> O	1.03	0.97	1.15	0.93	1.75	1.67	1.74
MoO <sub>3</sub>	<0.2	<b>5.16</b>	<0.2	<0.2	<0.2	<b>10.6</b>	<0.2
V <sub>2</sub> O <sub>5</sub>	7.18	<u>4.13</u>	<u>3.86</u>	<u>3.66</u>	20.7	<u>13.5</u>	<u>10.2</u>
NiO	1.17	0.42	1.17	1.04	2.68	1.94	1.91

Slags processed with a Ni crucible have an Fe and V deficit, but Ni content is almost unchanged from the initial P1 composition.

Figure 6.13a shows that viscosities of P1 slag processed in Ni were higher than those for P1 slag measured at the same temperatures in Al<sub>2</sub>O<sub>3</sub> and Mo crucibles. The measured viscosity values were obtained before melting of the Ni crucible. At these temperatures, slag is mainly liquid with approximately 6-7 wt.% V-rich solids. It is likely that viscosity of the slag in Ni was affected by formation of metallic Fe in the slag which would otherwise reduce viscosity as FeO. The impact of Mo on the viscosity is unknown, while V in the slag typically reduces viscosity of the slag above the liquidus temperature, working as network modifier [7]. However, at lower temperatures, the presence of V leads to formation of V-rich solids (e.g. spinel) and a significant increase of viscosity.

Therefore, a decrease of V content in P1 slag likely reduces the amount of V-rich solids formed in this system and does not lead to a sharp increase in viscosity at temperatures below 1300 °C as is observed for P1 slag processed in Al<sub>2</sub>O<sub>3</sub> [7]. The viscosity versus temperature trend for this slag almost follows a simple exponential dependency on temperature through the entire range of measurements (1200-1500 °C).

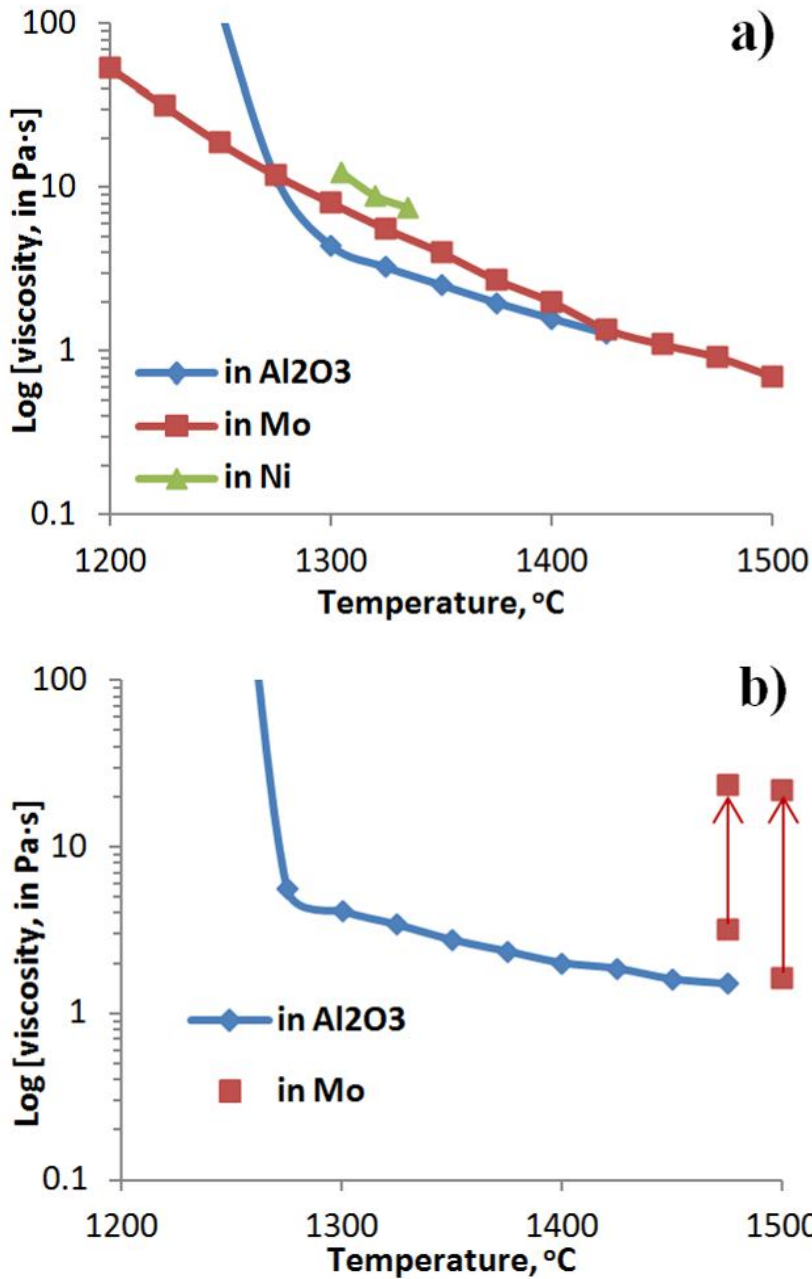


Figure 6.13: Viscosity versus temperature for P1 (a) and P2 (b) slags processed with different support materials.

Of the three crucible materials tested with P1,  $\text{Al}_2\text{O}_3$  provided the lowest viscosities in the temperature range 1300-1400 °C. It is likely that at temperatures where viscosity is mainly affected by composition of the liquid phase (above or near the liquidus temperature), the extraction of V from the P1 slag is partially or fully compensated by  $\text{Al}_2\text{O}_3$  enrichment from interaction with the crucible. Increased  $\text{Al}_2\text{O}_3$  content in the liquid lowers the Si/Al ratio, which likely lowers the viscosity for these particular conditions. However, at lower temperatures (<1300 °C) the excess of  $\text{Al}_2\text{O}_3$  leads to formation of spinel and a sharp increase of the viscosity.

For P2, slag viscosity measurements were conducted in Mo and  $\text{Al}_2\text{O}_3$  crucibles only (Figure 6.13b). For the slag processed in an  $\text{Al}_2\text{O}_3$  crucible, viscosity values were obtained for an extended temperature range. The measurements with a Mo crucible were unstable since P2 slag was highly reactive with Mo and extensive reactions affected the viscosity measurements through the entire testing procedure. The viscosity of the slag in a Mo crucible was non-Newtonian and also significantly varied over time. For P2 processed in  $\text{Al}_2\text{O}_3$ , the viscosity vs. temperature trend is very similar to that of P1 slag in  $\text{Al}_2\text{O}_3$ : viscosity values are low at high temperatures and follow an exponential dependence, and then a sharp increase of viscosity is observed at temperatures below 1270 °C. Similar to P1 slag in  $\text{Al}_2\text{O}_3$ , this viscosity behaviour is also affected by changes in  $\text{V}_2\text{O}_5$  and  $\text{Al}_2\text{O}_3$  concentrations in the slag.

It has to be noted that spindle materials (Mo and  $\text{Al}_2\text{O}_3$ ) contacts with slag and also contribute to slag chemistry changes. Post-viscosity measurement observations of the spindles do not indicate significant changes in the geometry of the spindles and their impact on viscosity value (less than 5%) does not exceed standard deviation of the measurements.

#### **6.4.6 Application of the results to viscosity measurements**

##### *6.4.6.1 Phase equilibrium approach*

Phase equilibrium considerations are used to predict flow behavior of metallurgical and coal gasification slags [10-15]. Coal slags are often described by their four major

components,  $\text{SiO}_2\text{-Al}_2\text{O}_3\text{-FeO}_x\text{-CaO}$ , and occasionally some minor components such as  $\text{MgO}$ ,  $\text{Na}_2\text{O}$ ,  $\text{K}_2\text{O}$ ,  $\text{TiO}_2$ . Hence, it is often desired to attain phase equilibrium prior to measuring slag viscosity. This approach avoids the added complexity of kinetics. It is appropriate for slags which are stable within the measuring system. The situation for petcoke slags is different and seeking phase equilibrium may not be the most appropriate approach.

Firstly, petcoke slags are highly reactive with the crucibles and the current work indicates continuous changes in the slag composition over 24 h of equilibration at a fixed temperature. It has been suggested that phase equilibria in V-rich slags can be achieved after 20-72 h of equilibration, depending on slag composition and temperature [8]. With such equilibration time, the bulk composition of the slags, processed with the crucibles and under conditions specified in the present work, definitely changes significantly and therefore the measured viscosity after equilibration would not be relevant to the initial petcoke composition.

Secondly, the slag residence time in an entrained flow gasifier may not allow full equilibration. The residence time of slag in an entrained flow gasifier depends on gasifier dimensions and slag viscosity which are affected by slag composition and temperature near the gasifier wall. Taking into account the dimensions of the Siemens 5 MWth gasifier [16], viscosity values acceptable for continuous slag flow (5-25 Pa·s) [17], and the assumption that slag flow is continuous with steady tapping rates, typical residence times were calculated as follows. For a given slag viscosity, slag velocity can be calculated using a simplified fluid flow equation which is derived from momentum balance equations applying Newton's law of viscosity [18]. The average velocity over a cross section of a flowing film can be calculated using equation 1:

$$v = \frac{\rho g \delta^2 \cos \beta}{3\mu} \quad (1)$$

where

v = average velocity of the slag, m/s

$\rho$  = density of the slag, kg/m<sup>3</sup> (a constant value of 2500 kg/m<sup>3</sup> in this calculation)

$g$  = acceleration due to gravity, 9.81 m/s<sup>2</sup>

$\beta$  = inclination angle (relative to a vertical orientation, e.g.,  $\beta = 0$  for a vertical wall)

$\mu$  = viscosity of the slag (Pa·s)

$\delta$  in equation 1 is the thickness of the film (m) which can be calculated using an equation proposed by Bird et al [18] as follows:

$$\delta = \sqrt[3]{\frac{3\mu w}{\rho^2 g W \cos\beta}} \quad (2)$$

where

$W$  = circumference of inner wall for vertical sections of reactor or average circumference of inner wall for inclined sections (m)

$w$  = slag mass flow (kg/s)

The slag mass flow is calculated by assuming all of ash content in coal becomes slag on the gasifier inner walls. Based on these calculations, slag residence time is estimated at 10-30 min for an optimal viscosity range of 5-25 Pa·s.

Hence, even when flowing slag is formed during coal gasification, phase equilibrium may not be fully achieved. This was demonstrated in a previous study by the analysis of phase compositions in tapped slags [19]. For flowing petcoke (or coal-petcoke blend) slags, phase equilibria would not be achieved within 10-30 min and interaction with an immobile slag layer on the gasifier wall or with the refractory material can occur. Therefore, the kinetics of slag transformation cannot be ignored. From this perspective, equilibrium viscosity measurements may not be representative of such industrial processes.

#### 6.4.6.2 Material selection

Considering the four materials investigated in this work, it is evident that there is no ideal slag crucible to be used for the viscosity measurements of petcoke slags.

Crucibles made of metals such as Pt and Ni can be used in viscosity measurements only in less reducing conditions than relevant to the present study to avoid metallic Fe and Ni formation. No metallic Fe precipitation and dissolution of Fe in Pt was observed when synthetic petcoke slags were processed in Pt envelopes at 1500 °C and  $O_2$  activity =  $10^{-8}$  [8]. An additional limitation for Ni crucibles is temperature. Temperature has to be below Ni's melting point (1455 °C) or even lower if Ni dissolves some of the metals from the slag.

Mo is a reasonable crucible material for slags containing less than 5 wt.%  $V_2O_3$ . It was demonstrated that for these slags, the contamination of Mo and loss of V do not significantly affect viscosity. For slag containing 5-10 wt.%  $V_2O_3$ , the measurements are possible but the results have to be considered in context of the compositional changes. With higher V concentrations, Mo should not be used for viscosity measurements. However, in practice most petcoke are blended with coal in gasification. Since the ash content of petcoke is significantly lower than the ash content of coal, it is likely that the  $V_2O_3$  of most coal-petcoke blend ashes will be below 10 wt.%.

$Al_2O_3$  has an advantage over other materials since it is usually a major component of petcoke ashes. Therefore, the contamination from the crucible only changes a balance of base elements in the petcoke system. V losses to the crucible are significant but about half of the initial concentration has been conserved during viscosity tests conducted in this study. One option to perform viscosity measurements with high V content melts is to design artificial petcoke ashes with excess V and a deficit in Al. The excess V and Al deficit would compensate the changes in Al and V concentrations during slag viscosity measurements. This approach can be taken for research requiring phase equilibria. The current study indicates that Al-V spinel forms an interface between the slag and  $Al_2O_3$  crucible. This interface decreases migration of the slag to the crucible, and  $Al_2O_3$  to the slag liquid phase. Therefore, the slags with excess V will eventually be equilibrated with

a spinel primary phase field, while the rest of the slag may contain the liquid phase in equilibria with other solid phases (eg. other V-rich phases and/or mullite). For this equilibrated system the viscosity can be measured at a specified temperature for only one specified bulk slag composition which consists of the liquid slag and solids in the liquid slag (i.e. not including the spinel interface). With temperature changes, the equilibrium between the interface and the rest of the slag will change. As a result, the composition of the liquid and solids within the slag, and therefore the bulk slag composition (excluding the interface) will change accordingly.

In entrained flow gasification, the reaction of V-containing slags with immobile slag on the gasifier wall (for water-cooled walls) or with gasifier refractory [20] has to be considered. To assess petcoke slag flow in industrial processes, a crucible material which reproduces the contacting material in the process may be most suitable. This is a potential experimental avenue to be explored.

## 6.5 Conclusion

Petcoke slags containing 6-21 wt.%  $V_2O_5$  were found to be highly reactive with the crucible materials used in this study. The interaction of slag with the crucibles always involves the transfer of slag constituents to the crucible. For Mo and  $Al_2O_3$  crucibles, the interactions also include dissolution of the crucible material into the slag.

Slag equilibration at temperatures just below the liquidus temperature of slags indicates that the investigated compositions are located in a V-rich oxide primary phase field. Interactions of slag and crucible materials lead to changes in bulk slag composition and appearance of new solid phases. An interface layer between the slag and crucible was formed and identified as V-rich oxide, Mo-V-oxide and spinel for slags processed with Ni, Mo and  $Al_2O_3$  respectively. The interface layer can act as a barrier for migration of the slag components to the crucibles and elements of the crucibles to the slags.

Compositional changes in the slags have an impact on viscosity and lead to differences in viscosities of slags produced with different crucibles.

The results of this study indicate that for slags generated during petcoke gasification, kinetics of the slag formation and slag interactions with immobile slag or refractory have to be considered.

Conditions for use of a particular crucible material for slag viscosity measurements are specified as:

- Ni and Pt crucibles can be used for viscosity tests at low reducing conditions where formation of metallic Fe and Ni can be minimised or avoided;
- Mo supports can be used for viscosity measurements where the total concentration of  $V_2O_3$  in the petcoke (or coal-petcoke blend) ash is less than 5 wt.%;
- $Al_2O_3$  crucibles can be used for slags with higher than 5 wt.%  $V_2O_3$ . However, the starting mixture has to be modified in order to compensate for changes in Al and V concentrations in the slag during viscosity tests.

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## **Chapter 7. Fate of inorganic matter in entrained-flow slagging gasifiers: Fuel characterization**

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## 7.1 Abstract

This study is the first of a three-part research program which involves fuel characterization, testing in a 1 MW<sub>th</sub> gasifier, and computational fluid dynamics (CFD) modeling for entrained-flow slagging gasification. Focus is on the behaviour of inorganic fuel components since the end goal is to develop a CFD model which includes inorganic matter transformations. Initially, four coals were selected for this program and one limestone was also chosen to act as a fluxing agent. Fuel properties related to ash particle formation, gas-particle transport, particle sticking, slag flow and slag-refractory interaction are provided with prioritization based on their potential application for screening of potential fuels, ensuring proper gasifier operation, gasifier design and/or CFD modeling. The selection of one or multiple experimental and/or modeling techniques is justified and applied to determine each relevant property. Of the four coals tested, one was deemed unsuitable based on initial screening tests. Two of the three remaining coals require fluxing for proper gasifier operation. Design tests showed that alumina is preferred over silicon carbide and alumina-chromia (with 30 wt.% chromia) for use as refractory material with the selected fuels. Characterization for CFD modeling is also discussed with results provided as supplementary data.

*Keywords:* Gasification, Inorganic, Mineral, Ash, Slag, CFD model

## 7.2 Introduction

Fuel and product flexibility have always been some of gasification's selling points. More recently, concerns over reducing emissions in industry and the power sector have renewed interest in this technology. In comparison with coal combustion systems, the advantages of coal gasification include high efficiency, suitability for carbon capture, relatively easy removal of sulphur oxides, nitrogen oxides and trace contaminants, and low water consumption [1]. Most of the successful high throughput coal gasifiers developed in the past 60 years are of the entrained-flow slagging type [2]. This study is the first part of a three-part research program which involves fuel characterization, testing in a 1 MW<sub>th</sub> gasifier, and computational fluid dynamics (CFD) modeling for entrained-flow slagging gasification. Here, the focus is on the behaviour of inorganic fuel components as this is still ill-understood even though it can be the determining factor in designing and operating entrained-flow gasifiers [3]. For instance, fouling or plugging of the gasifier by inorganic matter is a major concern. Refractory wear, which is largely dictated by interaction with inorganic matter, leads to poor plant availability. Also, the distribution of potentially toxic inorganic matter in fly ash and slag is an environmental concern. Four coals were selected for this program. One limestone was also selected to act as a fluxing agent to reduce slag viscosity if required for operation with any of the coals. F1 fuel is a lignite coal from Saskatchewan, Canada. F2 fuel is a sub-bituminous coal from Alberta, Canada. F3 fuel is a beneficiated version of F2 with reduced ash content. F4 fuel is another sub-bituminous coal from Alberta, Canada. L1 is a limestone from a Canadian power company. The characterization of the fuels is described in this study. Some bench-scale tests and models which do not relate to inorganic matter transformations are not included. For example, thermogravimetric analysis of the fuels can be used to optimize gasifier performance based on fuel reactivity, but is not directly related to inorganic matter transformations and is therefore not included.

Selection of fuel analyses depends on the intended application of the results: (i) screening of potential fuels; (ii) proper gasifier operation; (iii) gasifier design; and/or (iv) CFD modeling. Unless an accurate CFD model is readily available, fuel screening, gasifier operation and gasifier design will only make use of a subset of the tests required for CFD modeling. A comprehensive CFD model must consider all relevant inorganic matter phenomena, including ash particle formation, gas-particle transport, particle sticking, slag flow and slag-refractory interaction. The properties involved are listed in Table 7.1 along with their associated phenomena and experimental tests and models used to determine them. The likely applications of each property are given in the last column of Table 7.1. Since the end goal of this research program is to develop a CFD model which includes inorganic matter transformations, a test or model for each property must be used. In Table 7.1, experimental and modeling techniques underlined and in red font were applied in this study. Although modeling techniques generally require less time and resources, experimental techniques are preferred where model accuracies are poor and/or the property is very important for gasifier design and/or operation (e.g., slag viscosity). The reasoning for the selection of each technique or model used is included in the discussion. Properties related to screening were determined first, followed by the properties related to operation, design, and finally CFD modeling. Results pertaining to screening, operation and design are supplied directly in the paper. For brevity's sake, results which only pertain to CFD modeling are provided as supplementary information. However, the methods and importance of all results are discussed with references which demonstrate the use of each property in CFD modeling.

**Table 7.1: Fuel properties involved in inorganic matter phenomena**

Property	Phenomena <sup>a</sup>	Experimental	Model	Applications <sup>b</sup>
fuel composition	1	<u>ultimate analysis</u> , <u>proximate analysis</u> , <u>calorimetry</u>	N/A	S, O, D, M
fuel form	1	<u>coal petrography</u> , CCSEM	ultimate/proximate models	M
fuel particle size	1, 2, 3	<u>sieving</u> , <u>laser diffraction</u> , CCSEM	N/A	M
inorganic composition	1, 2, 3, 4, 5	<u>XRF</u> , XRD, CCSEM	N/A	S, O, D, M
inorganic form	1	<u>XRD</u> , HT-XRD	<u>FactSage</u>	M
inorganic particle size	1, 2, 3	<u>CCSEM</u>	N/A	M
fuel-inorganic associations	1	<u>CCSEM</u> , chemical fractionation	N/A	M
ash melting temperatures	3, 4	<u>AFT</u>	composition models, <u>FactSage</u>	S, O, D, M
slag viscosity	3, 4, 5	<u>rheology</u>	<u>composition models</u>	O, D, M
slag-refractory reactivity	4, 5	<u>cup tests</u> , sessile drop	<u>FactSage</u> , numerical diffusion models	D, M
slag interfacial tension	3, 5	sessile drop	<u>composition models</u>	M
inorganic density	1, 2, 3, 4, 5	pycnometry, buoyancy, dilatometer, sessile drop	<u>composition models</u>	M
inorganic heat capacity	1, 4, 5	calorimetry	<u>composition models</u> , FactSage	M
inorganic heat conductivity	1, 4, 5	various techniques with heating and temperature measurements	temperature models, <u>thermal diffusivity models</u>	M
inorganic emissivity	1, 4	FTIR	<u>constant value from literature</u>	M

a) Phenomena are coded as (1) ash particle formation, (2) gas-particle transport, (3) particle sticking, (4) slag flow and (5) slag-refractory interaction.

b) Applications are coded as (S) screening, (O) operation, (D) design and (M) CFD modeling.

## 7.3 Experimental

### 7.3.1 *FactSage modeling*

FactSage software predicts equilibrium solid-liquid-gas phases and compositions based on Gibbs free energy minimization [4]. Gibbs free energy is calculated from optimized models with parameters based on empirical data with various compositions, temperatures and pressures. This information is contained within compound and solution databases which must be carefully selected prior to equilibrium calculations. The FactSage 6.2 *Equilib* module was utilized with the FACT53 and FToxid databases. Oxide components representing less than 1 wt.% for all samples were excluded. Sulphur is not included in the analyses as it is expected to completely devolatilize at the temperatures of interest. All gas and solid compounds were considered. For compounds found in both the FACT53 and FToxid databases, preference was given to the FToxid database to have better thermodynamic consistency with the solution phases data. Pure liquid Fe was also considered. All solution phases were considered. However, only the SLAGA solution was allowed to form multiple immiscible phases. Total pressure was specified at 1 atm and, unless otherwise noted, the activity of O<sub>2</sub> in the gas phase was set to 10<sup>-9</sup> which is similar to what is typically encountered near the walls in a gasifier [5].

### 7.3.2 *Slag viscosity measurements*

Artificial ashes were prepared by mixing laboratory or analytical grade Al<sub>2</sub>O<sub>3</sub>, CaCO<sub>3</sub>, Fe<sub>2</sub>O<sub>3</sub>, K<sub>2</sub>CO<sub>3</sub>, MgCO<sub>3</sub>, Na<sub>2</sub>CO<sub>3</sub>, SiO<sub>2</sub> and TiO<sub>2</sub> powders. Slag viscosity measurements were performed in a Carbolite 3 litre BLF1700 furnace with a Brookfield RVDV-IIIU rheometer. Both the furnace and the rheometer were connected to a personal computer for external control and data collection. Samples were placed in a molybdenum crucible. The crucible was placed in a custom-built alumina chamber within the furnace for gas atmosphere control and spill protection. A type B thermocouple was inserted into the

chamber to monitor the sample temperature. Argon flowed at 0.4 dm<sup>3</sup>/min (STP) into the alumina chamber and a graphite sleeve was placed around the molybdenum crucible to consume trace oxygen. Mössbauer spectroscopy of a slag sample after viscosity measurements indicated that the iron components were in the ferrous state, confirming the reducing nature of the experimental conditions. A molybdenum spindle was attached to the rheometer and lowered until it sufficiently penetrated into the slag sample for the purpose of accurate viscosity measurement. It was rotated while measuring the torque required for spindle rotation at a given angular velocity. The operation of the rheometer was validated with the National Institute of Standards and Technology's (NIST, United States) standard reference material 717a. Artificial ash samples were first heated to 1225 °C and left overnight for equilibration. They were then heated to 1525-1600 °C. Measurements were conducted at incremental temperature reduction steps. The cooling step for each set of measurements included a reduction in temperature of 25 °C with a minimum of 25 minute intervals to allow the temperatures and compositions to equilibrate inside the crucible. At each temperature, viscosity was measured multiple times at various rotational speeds to detect non-Newtonian behaviour [6]. The final measured value was taken at approximately 80-90% of the rheometer's maximum torque.

### ***7.3.3 Cup tests for slag-refractory reactivity***

Approximately 0.5 g of ash was placed on 1200 mm<sup>2</sup> pieces of refractory. The crucible was placed in a custom-built alumina chamber within a Carbolite 3 litre BLF1700 furnace for gas atmosphere control and spill protection. A type B thermocouple was inserted into the chamber to monitor the sample temperature. Argon was fed at 0.4 dm<sup>3</sup>/min and small pieces of graphite were placed near the samples to consume any trace oxygen. Mössbauer spectroscopy of a slag sample after testing indicated that all the iron atoms are in the ferrous state, confirming the reducing nature of the experimental conditions. Samples were either heated to 1250 or 1500 °C at approximately 3 °C/min. They were then held at the temperature of interest for 4 h and afterwards, the samples were cooled at approximately 5 °C/min to 1100 °C and then slowly cooled to room temperature. The slag-covered refractory was cross-sectioned using a Buhler ISOMET

low speed saw with a 0.4 mm 15LC wafering blade. The samples were mounted with carbon tape on an aluminum holder. Scanning electron microscopy with energy-dispersive X-ray spectroscopy (SEM-EDX) analysis was done using a Hitachi 3400N VP-SEM with Oxford Instruments Si(Li) Pentafet Plus 10 mm<sup>2</sup> detector. The system was operated at a reduced vacuum of 30 Pa to reduce charging from the uncoated samples. Imaging and analyses were done at 20 kV and approximately 60  $\mu$ A emission current. X-ray acquisition was done with Oxford INCA *Point and ID* and *Mapping* modules.

#### **7.3.4 Coal petrography to determine fuel form**

Coal samples (< 0.85 mm) were made into polished 25.4 mm diameter coal pellets for microscopical analysis by reflected light as per ASTM D 2797-11a. Petrographic analysis included (i) determination of v-type groups, percent reflectance of vitrinite and Romax as per ASTM D 2798-11a and (ii) determination of the volume percent of macerals or coalified organic remains as per ASTM D 2799-05a. Vitrinite reflectance was performed with a Zeiss microscope photometer equipped with a polarizer using a 40x oil immersion objective, 12.5x oculars, a tube factor of 1.6 and total magnification of 800x. In this work 100 readings on vitrinite macerals covering the entire pellet surface were recorded. Identification and point counting of coal macerals was done using a Leitz S 544-21 microscope; 50x oil immersion objective, 10x oculars, tube factor of 1.6, total magnification of 800x. 1000 maceral counts were made on the entire polished pellet surface.

#### **7.3.5 CCSEM**

Computer-controlled scanning electron microscopy (CCSEM) samples were mounted in carnauba wax and carbon coated. The wax mount was glued to an epoxy blank for structural strength. Analysis was done using a Hitachi 3400N VP-SEM with Oxford Instruments Si(Li) Pentafet Plus 10 mm<sup>2</sup> detector. The system was operated at full vacuum. Imaging and analyses were done at 20 kV and approximately 60  $\mu$ A emission current. X-ray acquisition was done with Oxford INCA Feature software. Three analyses

were conducted per sample, all using the backscatter detector to provide elemental contrast in the image. The first was 10,000 particles acquired at 500x magnification. This collected particles over 10 pixels in size (0.89  $\mu\text{m}$  equivalent circle diameter) and under 5  $\mu\text{m}$  in breadth. A 200x magnification analysis was done over the same area as the 500x run to collect particles over 5  $\mu\text{m}$  but under 30  $\mu\text{m}$  breadth. Particles over 30  $\mu\text{m}$  breadth were collected over the same area at 75x magnification. Energy-dispersive X-ray spectroscopy (EDX) data was collected for 5 seconds per particle, at the center of the longest chord. Each hour the quantification optimization was performed with cobalt. Two particle detection grey level ranges were used and set manually before each analysis. For the limestone sample, only 1 grey level range was used. Coal particles were kept separate from mineral particles during the automated detection. Remove edge touching, two erode, one dilate and hole fill filter options were used in the image processing to help separate particles that were close to each other. All three magnifications were combined and the data was sorted. Coal particles were identified by grey-level analysis; all particles below a grey level determined at the start of the analysis were classed as coal. The remaining particles were classified by their elemental composition. C was excluded due to the high background, and O was calculated using the O by stoichiometry option. Quantification was done with the base Oxford factors, no additional standards were used.

### **7.3.6 Other methods**

ASTM D7582 was applied for determination of moisture, secondary moisture, ash and fixed carbon in the proximate analyses. ISO 562 was applied for determination of volatile matter in the proximate analyses. ASTM D5373 was applied for determination of carbon, hydrogen and nitrogen in the ultimate analyses. ASTM D4239 was applied for determination of sulphur in the ultimate analyses, while oxygen was calculated by difference. ISO 1928 was applied to determine the gross calorific value of samples. ASTM D1857 was applied for determination of ash fusion temperatures. ASTM D4326 was applied to determine major and minor oxides content via X-ray fluorescence (XRF). Particle size analysis was conducted using a combination of sieving (grains > 2 mm),

Camsizer (grains 0.063-2 mm) and a Lecotrac LT100 Particle Size Analyzer (grains < 0.063 mm).

Quantitative X-ray diffraction (XRD) analysis of the samples was performed over the angular range of 7 to 80° (2 $\theta$ ). Each sample was mixed with an internal standard. The resulting mixture was thoroughly ground using a mortar and pestle to ensure sample homogeneity. Integrated intensities for strong diffracting peaks of each phase were obtained using JADE version 7.0 XRD processing software. Concentrations of the chemicals identified in each sample were calculated using the reference intensity ratio (RIR) method.

## **7.4 Results and Discussion**

### **7.4.1 Screening**

When multiple fuels are available and the gasification system in question need not rely on all of them, routine standardized tests and simple models can be applied to the fuels for screening. This step can reduce the number of fuels to be examined in the often more difficult and expensive tests and models for gasifier operation, design and CFD modeling. In this study, fuel composition, inorganic composition and ash melting temperatures have been selected for fuel screening.

Fuel composition is directly involved in only one inorganic matter phenomenon, ash particle formation. However, it is an indicator of two important fuel selection criteria, energy content and reactivity. In terms of gasifier operation, the fuel composition is required to determine fuel and fluxant mixing ratios. In terms of gasifier design, emissions can be estimated from the fuel composition and this will impact downstream processing. Also, CFD models rely on fuel composition for mass and energy balances. The proximate analysis, ultimate analysis and gross calorific value for the fuels in this

study are given in Table 7.2. The volatiles content from the proximate analysis can be correlated to reactivity [7]. The gross calorific value provides the energy content.

Inorganic matter composition is directly involved in all the inorganic matter phenomena. This property is an input to models for all properties which can be determined by FactSage or composition models (see Table 7.1) including the ash melting temperatures which are used for fuel screening. There are different options to experimentally determine inorganic matter composition. Quantitative XRD can be used for crystalline materials, but it fails to determine the composition of amorphous components. CCSEM is an option, but it is not a standardized technique. Hence, XRF was selected in this study and the results are presented in Table 7.3. It should be noted that the bulk inorganic matter composition is sufficient for screening, design and operation, but the composition of specific particles may be preferred for CFD modeling.

**Table 7.2: Proximate analysis, ultimate analysis and gross calorific value**

	<b>F1</b>	<b>F2</b>	<b>F3</b>	<b>F4</b>
<b>Proximate Analysis</b>				
Moisture (wt.%)	6.43	2.86	3.32	6.08
Ash (wt.%)	13.87	32.84	10.52	24.72
Volatile (wt.%)	36.27	27.30	34.76	26.87
Fixed Carbon (wt.%)	43.43	37.00	51.40	42.33
<b>Ultimate Analysis</b>				
Carbon (wt.%)	57.7	48.8	67.8	52.2
Hydrogen (wt.%)	3.64	3.00	4.22	3.05
Nitrogen (wt.%)	1.02	0.64	0.86	0.79
Sulphur (wt.%)	0.93	0.18	0.27	0.30
Oxygen (wt.%)	16.41	11.67	13.01	12.85
Gross Calorific Value (MJ/kg)	22.15	18.83	26.67	19.98

**Table 7.3: Inorganic matter composition**

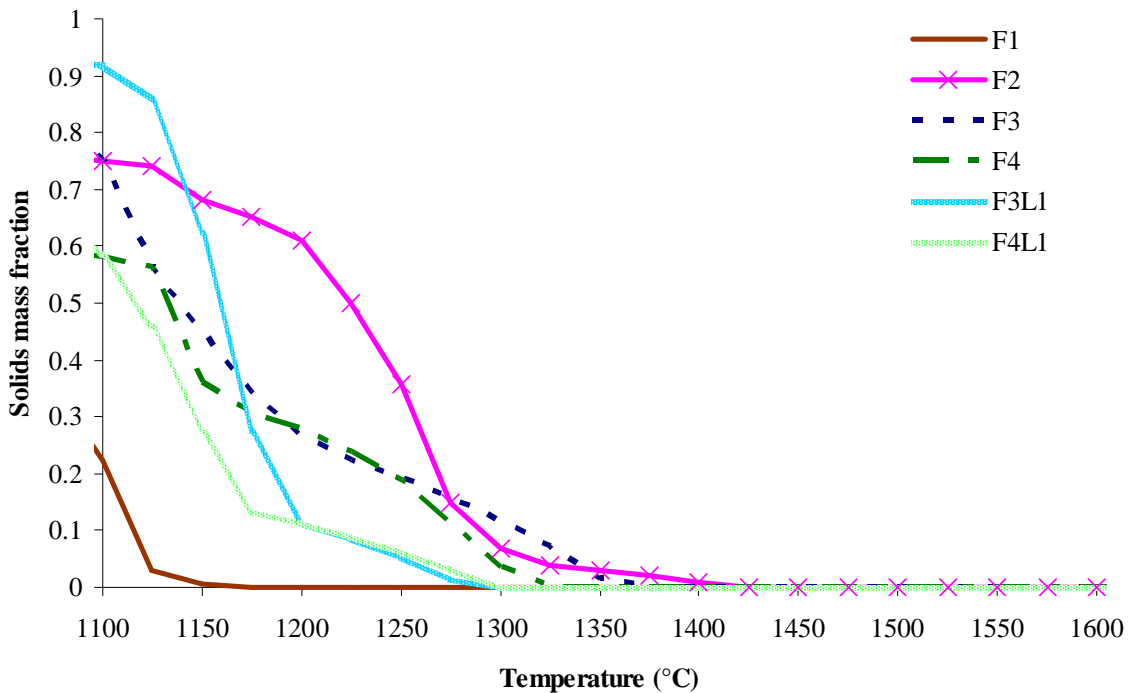
	<b>F1</b>	<b>F2</b>	<b>F3</b>	<b>F4</b>	<b>L1</b>
SiO <sub>2</sub> (wt.%)	37.97	67.32	53.47	57.07	1.15
Al <sub>2</sub> O <sub>3</sub> (wt.%)	16.31	17.28	17.7	19.42	0.18
Fe <sub>2</sub> O <sub>3</sub> (wt.%)	6.29	4.37	6.6	4.59	0.1
TiO <sub>2</sub> (wt.%)	0.71	0.59	0.62	0.72	<0.03
P <sub>2</sub> O <sub>5</sub> (wt.%)	0.39	0.14	0.57	0.06	<0.03
CaO (wt.%)	13.56	4.19	11.25	6.22	48.45
MgO (wt.%)	2.73	1.29	1.75	1.44	0.56
SO <sub>3</sub> (wt.%)	14.29	1.12	4.68	5.06	<0.10
Na <sub>2</sub> O (wt.%)	6	0.39	0.4	2.34	<0.20
K <sub>2</sub> O (wt.%)	0.27	2.26	1.22	1.7	0.09
BaO (ppm)	4478	1641	3781	2642	<250
SrO (ppm)	3572	295	837	977	147
V <sub>2</sub> O <sub>5</sub> (ppm)	56	97	135	140	<50
NiO (ppm)	<50	<50	50	71	<50
MnO (ppm)	253	424	829	324	<50
Cr <sub>2</sub> O <sub>3</sub> (ppm)	<50	51	<50	101	<50
CuO (ppm)	48	<30	48	65	<30
ZnO (ppm)	39	84	97	94	<30
Loss on Fusion (wt.%)	0.63	0.79	1.18	0.93	49.44
Sum (wt.%)	100	99.99	99.99	100	99.97

Ash melting temperatures are very important for ash particle formation, ash particle stickiness and slag flow. Li *et al.* performed a series of experiments with partially converted char particles in a laminar entrained flow reactor under gasification conditions [8,9]. Char particles would only stick if they were above a certain carbon conversion (~90% for Illinois #6 coal) and at a temperature above the ash flow temperature. For a slag layer formation model, Yong assumed sticking when both the particle and surface are liquid, and rebounding when both are solid [10]. When either the particle or surface is solid and the other component is liquid, the Weber number was applied. Rebounding was predicted when this number exceeded 1, otherwise sticking was predicted. To model slag flow, knowing the melting temperatures is critical as partially solidified slag tends to

have a much higher viscosity than fully molten slag and may display non-Newtonian behaviour [6], while fully solidified slag will not flow at all. In terms of fuel screening, a high melting temperature indicates a poor candidate for gasification. However, low melting temperatures do not guarantee the fuel's slag will flow well, since the viscosity may still be high. Slag viscosity tests may be used for screening, but slag viscosity measurements are difficult and existing models are often unreliable [11]. The melting temperatures can impact operation as temperature in the gasifier may be controlled to specifically avoid solidification. They may also affect design of the gasifier as it may impact reactivity with the refractory material. Ash fusion temperatures (AFTs) are determined by ashing a fuel via a standard procedure, forming a cone, pyramid or cube with the ash, and observing the deformation of the molded ash as it is heated up. The procedure is widely employed, standardized and inexpensive. AFTs are used as indicators of degrees of melting for the ash at different temperatures, and it is often assumed that the ash is fully molten at the highest AFT (the flow temperature). However, solids may be present at all fusion temperatures provided by the standard methods. Also, some ash minerals may melt at a given temperature, then react with other molten minerals at a higher temperature and form solids [12]. Factors other than liquid fraction, such as surface tension, will affect the shape of the mold. Since AFTs are determined upon heating of the ash, they are most applicable to ash particle formation and stickiness. AFTs for this study's fuels are given in Table 7.4. Reactor plugging usually happens with cooling of slag as it contacts the refractory and flows down towards the cooler slag tap. Solidification behaviour can be very different from melting [7]. If equilibrium can be assumed during cooling, FactSage predictions are appropriate. However, these are subject to modeling error. The mass fraction of solids in the slags of each fuel, as predicted using FactSage, are plotted in Figure 7.1.

**Table 7.4: Ash fusion temperatures in °C**

	F1	F2	F3	F4
<b>Oxidizing</b>				
Initial	1191	1254	1238	1196
Spherical	1268	1332	1274	1279
Hemispherical	1299	1385	1313	1327
Fluid	1332	>1510	1416	1457
<b>Reducing</b>				
Initial	1096	1213	1177	1191
Spherical	1107	1293	1263	1246
Hemispherical	1118	1393	1324	1310
Fluid	1177	>1510	1374	1446



**Figure 7.1: Solids mass fractions predicted using FactSage.**

F1 and F3 have higher volatile content which may result in better reactivity, higher gross calorific values which would provide greater efficiency, and lower ash content which reduces the risk of fouling and plugging. Their inorganic matter contains less SiO<sub>2</sub> and

more CaO, which is favorable to reduce slag viscosity [6]. The lower AFTs also favor F1 and F3. F1 has the lowest solids mass fractions, followed by F3 and F4 which have similar fractions, and F2 which has the highest. Nonetheless, limestone may be added to the fuels to improve their ash melting temperatures and slag flow properties. However, F2 will likely require more limestone than the other coals. This may be problematic as it already has very high ash content. Therefore, F2 was not evaluated for operation, design, and CFD modeling.

### **7.4.2 Operation**

Certain variables can be adjusted while operating a gasifier, such as the fuel blend, fuel feeding rate, oxidant feeding rate and moderator (i.e., steam) feeding rate. Adjusting the feeding rates will change the temperature and gas composition within the reactor, while changing the fuel blend can change the temperature required for proper slag flow. In this study, slag viscosity and properties used for screening were applied for proper gasifier operation.

Slag viscosity plays a role in particle sticking, slag flow and slag-refractory interaction. One approach in estimating the particle sticking probability involves calculating the kinetic energy of the impacting particle and the energy dissipation at impact [13,14]. If the particle has more kinetic energy than the amount which can be dissipated by the collision, it is assumed that the particle rebounds. Otherwise, it is assumed that the particle sticks. Another approach to predicting the sticking probability is to use a viscosity criterion [15,16]. Both approaches require knowledge of slag viscosity. For slag-refractory interaction, viscosity of the slag controls the dissolution rate of refractory and slag penetration into the refractory [17,18]. Viscosity is also a critical property in slag flow modeling [19,20].

In this study it was assumed that the temperature near the slag tap would be 1400 °C. As a rule of thumb, the viscosity should be below 25 Pa·s at the slag tap in order to maintain proper slag flow [6,21-23]. Slag with higher viscosity will likely plug the reactor. To

provide some leeway, a viscosity of 10 Pa·s or less at 1400 °C was the target for the fuel blends in this study. Measured slag viscosities for each fuel are given in Figure 7.2. The slag viscosity for F1 is well below 10 Pa·s at 1400 °C. However, slag viscosities for F3 and F4 do not meet the required specification. To minimize the number of measurements required to determine how much L1 should be added to the fuels, various models were relied upon. Using the SlagViscosityPredictor tool [11], predictions from several viscosity models were compared to the measured values. The model which performs best with each fuel was then used to predict viscosity at 1400 °C with various L1:fuel ratios. It was determined that the mass ratio of L1:F3 should be 0.042:1, and the mass ratio of L1:F4 should be 0.055:1. These optimized L1:F3 and L1:F4 blends are henceforth referred to as F3L1 and F4L1, respectively. Since the viscosity models perform poorly when solids are present in the slag, solids mass fractions were determined for the blends using FactSage (Figure 7.1). The limestone and fuel blends have lower solids than the unblended fuels above 1175 °C; therefore viscosity measurements were performed to confirm the viscosity predictions. Results are shown in Figure 7.2. In fact, the slag viscosities for the blends are indeed below 10 Pa·s at 1400 °C and therefore these blends are suitable for use in the gasifier. However, if the gasifier's refractory material reacts with the slag, the viscosity can be altered by mechanisms such as dissolution changing the liquid composition and spalling adding solids to the slag. Including these effects requires a more thorough investigation.

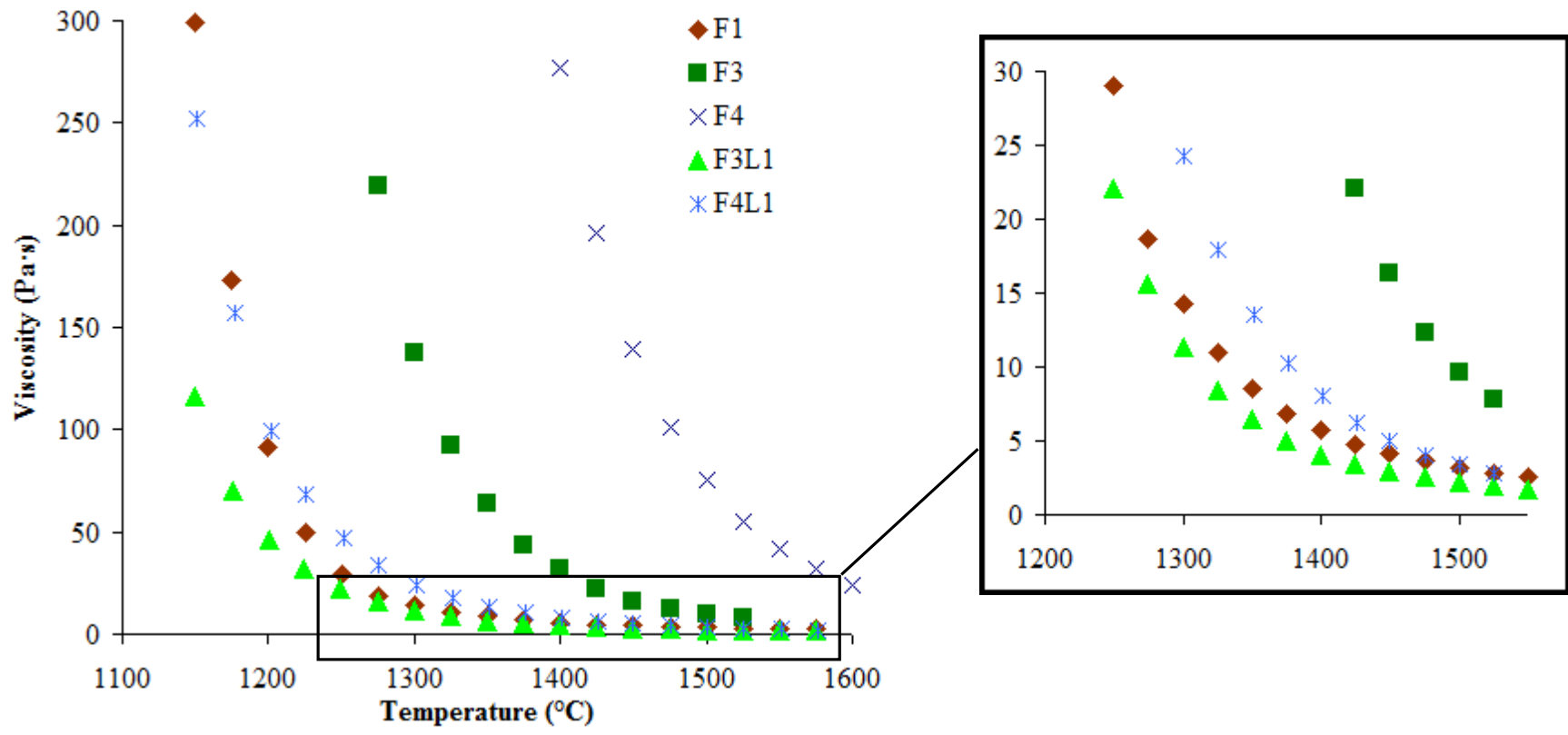


Figure 7.2: Measured slag viscosities.

### **7.4.3 Design**

Many variables will affect the design of an entrained-flow gasifier. Different options are available in terms of configurations (e.g., bottom-fired or top-fired), burner structure (e.g., with or without induced swirl), or the cooling mechanism (e.g., air-cooled or water-cooled) [2]. All of these choices will affect the fate of inorganic material, but the effects are difficult to predict without a comprehensive CFD model and the selection of these options is usually based on other factors. Nonetheless, there is freedom in selecting the type of refractory material. This design choice can significantly impact the fate of inorganic matter due to slag-refractory interaction. It will also affect the lifetime of the refractory liner which is a major cost for gasifier operation both due to replacement material cost and loss of production during replacement [24]. Refractories are usually alumina-based; sintered and/or fused cast alumina-silicate, high alumina, chromia-alumina, chrome-magnesia spinels or alumina-chromia-zirconia [5]. Silicon carbide can also be used but is more typical of water-cooled reactors [2]. In this study, alumina, silicon carbide and alumina-chromia (with 30 wt.% chromia) were considered. Slag-refractory reactivity and slag viscosity were used to predict the effect of refractory material on inorganic matter. FactSage modeling and cup tests were applied for slag-refractory reactivity predictions. Numerical models simulating compressive/tensile forces [25-27], which take effect when slag infiltrates refractory, were not applied since they require information which is not currently available for the gasification system in question (e.g., temperature gradient through the refractory which depends on the thickness of the slag layer).

Thermodynamic calculations with FactSage are often relied upon to model refractory corrosion by slag [17,18,28]. Typically, the difference between the concentration of refractory material in the slag saturated with refractory and its concentration in the original slag is assumed to be the driving force for corrosion by diffusion with the concentration in saturated slag determined using FactSage. In this study, Fick's first law of diffusion is combined with the Stokes-Einstein equation such that:

$$J \propto F = \frac{T}{\eta} \Delta C$$

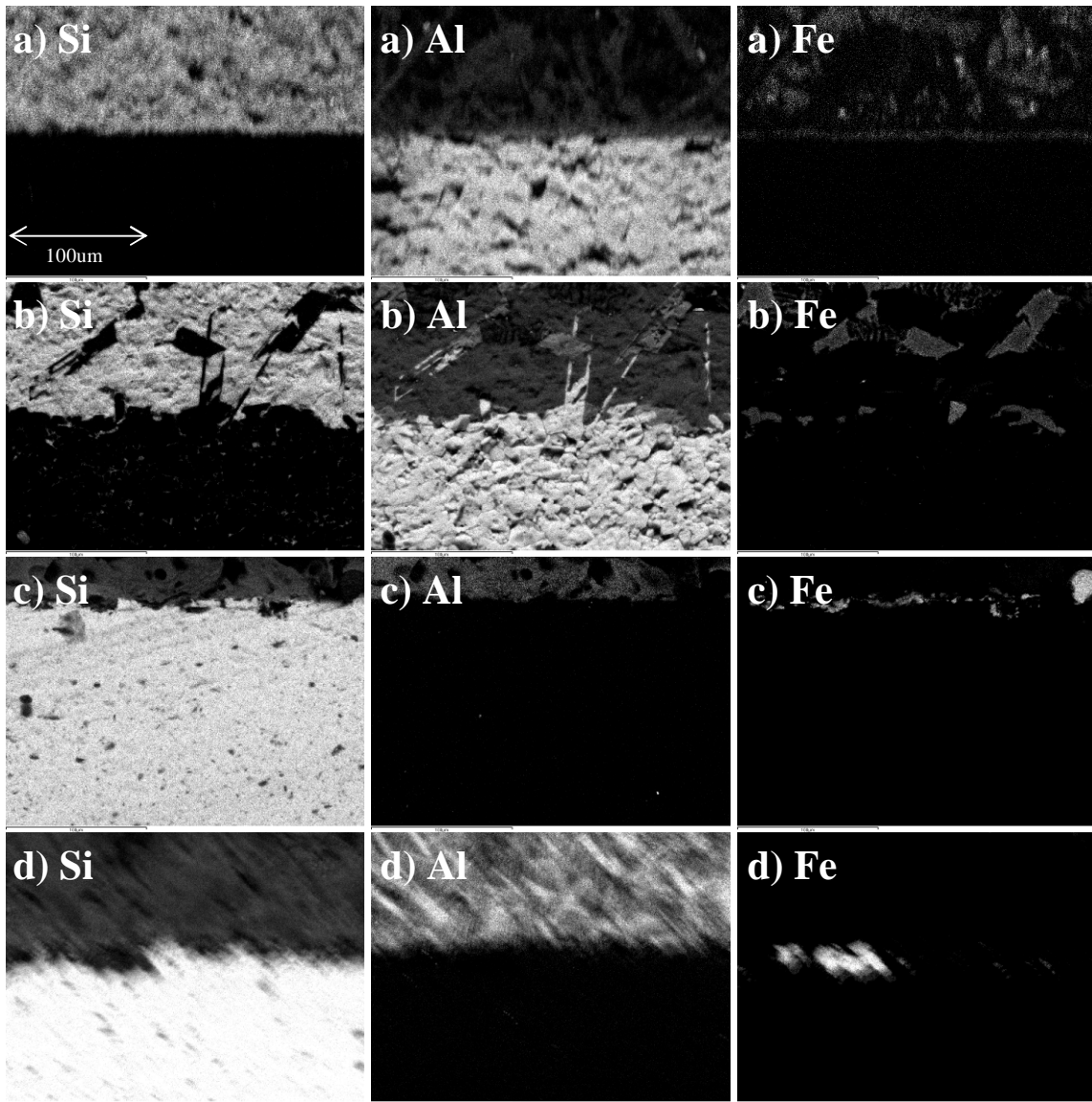
where  $J$  is the diffusion flux of refractory in slag,  $F$  is the diffusion factor,  $T$  is the absolute temperature,  $\eta$  is viscosity of the slag, and  $\Delta C$  is the concentration driving force. A greater diffusion factor indicates a higher diffusion flux and greater corrosion. To obtain a rough estimate of the diffusion factor, the following three assumptions are applied. i) The effect of the diffusing molecule on the slag's viscosity is negligible. ii) Molecules representing the average composition of the refractory are dissolved. Hence the slags will have an initial concentration of alumina which can be determined from the fuel ash composition (Table 7.3), excluding  $\text{SO}_3$  which devolatilizes at the temperatures of interest. However, they will have an initial concentration of 0 wt.% with respect to silicon carbide and alumina-chromia. iii) The slag is saturated with refractory when the refractory equilibrium solid phase is formed upon further addition of refractory to the slag. The equilibrium solid phase is the dominant phase when a large excess of refractory is added. This phase and the saturation concentration can be determined using FactSage. Table 7.5 provides the slag viscosity (taken from Figure 7.2), refractory equilibrium solid phase, other solids present with refractory saturation, the concentration driving force and diffusion factor for five combinations of fuel slag and refractory material at 1250 and 1500 °C. The parameters used for the FactSage predictions are as described in the experimental section. For predictions with silicon carbide refractory, the activity of oxygen gas was not specified at  $10^{-9}$  since it is believed that the activity will be lower near the silicon carbide surface. For cases listed in Table 7.5 where solids other than the refractory equilibrium solid phase are present at saturation, it may be possible that the kinetics for formation of these solids is very slow and would not be observed in gasifiers. This could lead to lower concentrations at which the refractory reaches saturation in the slag. For this reason, a metastable diffusion factor was calculated (Table 7.5) for which solids other than the refractory equilibrium solid were not permitted in the FactSage predictions. From the diffusion factors, it is expected that corrosion by diffusion is 6-12 times greater at 1500 °C than at 1250 °C for all slag-refractory combinations. With metastable conditions, corrosion is predicted to be 10-70 times greater at 1500 °C than at

1250 °C for all slag-refractory combinations. Corrosion resistance seems to be greater for silicon carbide than the other refractory materials. Alumina and alumina-chromia seem to have similar corrosion resistance in non-metastable conditions. However, alumina-chromia has much better resistance in metastable conditions.

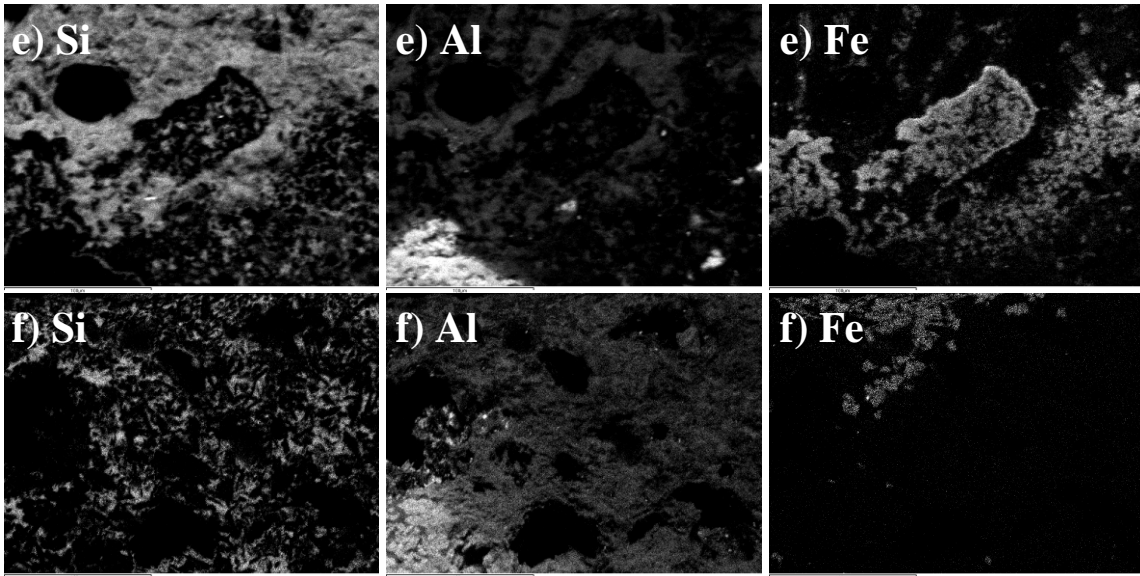
The approach to predicting corrosion described above only accounts for corrosion by diffusion and ignores other major corrosion mechanisms such as slag penetrating refractory and physical attributes of the refractory. Experimental results may provide the information lacking in a model-based approach. Much of the experimental data studying these mechanisms derive from cup tests [5,29] or sessile-drop type experiments [30,31]. Cup tests were performed for all the cases listed in Table 7.5. Refractories covered in slag were cross-sectioned for SEM-EDX analysis. Elemental maps are shown in Figure 7.3. Table 7.6 lists the composition, expressed as normalized atomic percentages, of the bulk slags after the cup tests. These were calculated by averaging 20 X-ray spectra for each slag. The initial slag composition, based on data in Table 7.3, is provided for comparison. Table 7.7 lists the composition, expressed as normalized atomic percentages, of the refractories after the cup tests. These were calculated for each refractory by averaging 20 X-ray spectra near the slag-refractory interface. The refractory compositions without slag, calculated by averaging 20 X-ray spectra, are provided for comparison. Note that C is detected by SEM-EDX, but it was not included in Tables 7.6 and 7.7 due to high background interference seen with this element.

**Table 7.5: Diffusion predictions**

Fuel slag	Refractory material	Temperature (°C)	Viscosity (Pa·s)	Refractory equilibrium solid phase	Other solids present at saturation	Concentration driving force (g/g)	Diffusion factor (K·Pa <sup>-1</sup> ·s <sup>-1</sup> )	Metastable concentration driving force (g/g)	Metastable diffusion factor (K·Pa <sup>-1</sup> ·s <sup>-1</sup> )
F1	alumina	1250	29.1	corundum	spinel, anorthite, Ca <sub>2</sub> Mg <sub>2</sub> Al <sub>28</sub> O <sub>46</sub> , CaMg <sub>2</sub> Al <sub>16</sub> O <sub>27</sub>	0.290	15	0.103	5
F1	alumina	1500	3.2	corundum	none	0.248	137	0.248	137
F3L1	alumina	1250	22.1	corundum	hibonite, spinel, anorthite, Ca <sub>2</sub> Mg <sub>2</sub> Al <sub>28</sub> O <sub>46</sub> , leucite	0.407	28	0.107	7
F3L1	alumina	1500	2.2	corundum	None	0.231	187	0.231	187
F4L1	alumina	1250	46.6	corundum	spinel, hibonite, anorthite, Ca <sub>2</sub> Mg <sub>2</sub> Al <sub>28</sub> O <sub>46</sub>	0.289	9	0.093	3
F4L1	alumina	1500	3.4	corundum	none	0.209	109	0.209	109
F1	silicon carbide	1250	29.1	silicon carbide	FeSi, graphite	0.065	3	0.020	1
F1	silicon carbide	1500	3.2	silicon carbide	FeSi, graphite	0.065	36	0.020	11
F1	alumina-chromia	1250	29.1	corundum	spinel, anorthite	0.248	13	0.010	1
F1	alumina-chromia	1500	3.2	corundum	spinel	0.242	134	0.065	36



**Figure 7.3: Silicon (Si), aluminum (Al) and iron (Fe) elemental maps produced by SEM-EDX of F1 on alumina at 1250°C (a), alumina at 1500°C (b), silicon carbide at 1250°C (c), silicon carbide at 1500°C (d), alumina-chromia at 1250°C (e) and alumina-chromia at 1500°C (f). The scale in the left (a) image applies to all images. (part 1 of 2)**



**Figure 7.3:** Silicon (Si), aluminum (Al) and iron (Fe) elemental maps produced by SEM-EDX of F1 on alumina at 1250°C (a), alumina at 1500°C (b), silicon carbide at 1250°C (c), silicon carbide at 1500°C (d), alumina-chromia at 1250°C (e) and alumina-chromia at 1500°C (f). The scale in the left (a) image applies to all images. (part 2 of 2)

**Table 7.6: Normalized atomic compositions of slags reacted with refractories**

	<b>Si</b>	<b>Al</b>	<b>Fe</b>	<b>Ca</b>	<b>Mg</b>	<b>Na</b>	<b>Cr</b>
<b>F1</b>							
initial	41.2	20.9	5.1	15.8	4.4	12.6	N/A
alumina 1250 °C	40.6	26.9	3.3	12.7	2.4	14.1	N/A
alumina 1500 °C	39.1	36.5	2.5	13.8	1.4	6.8	N/A
silicon carbide 1250 °C	N/A	N/A	N/A	N/A	N/A	N/A	N/A
silicon carbide 1500 °C	53.2	24.7	0.1	16.8	5.1	0.0	N/A
alumina-chromia 1250 °C	42.3	21.9	3.7	15.7	3.2	12.6	0.7
alumina-chormia 1500 °C	N/A	N/A	N/A	N/A	N/A	N/A	N/A
<b>F3L1</b>							
initial	46.3	18.0	4.3	28.2	2.5	0.7	N/A
alumina 1250 °C	46.9	19.7	4.0	26.1	2.5	0.9	N/A
alumina 1500 °C	30.3	41.0	3.7	22.2	1.5	1.3	N/A
<b>F4L1</b>							
initial	48.6	19.4	2.9	23.2	2.1	3.8	N/A
alumina 1250 °C	50.9	19.4	3.0	21.6	2.0	3.1	N/A
alumina 1500 °C	41.5	38.4	1.3	16.2	0.5	2.0	N/A

**Table 7.7: Normalized atomic compositions of refractories reacted with slag**

	<b>Si</b>	<b>Al</b>	<b>Fe</b>	<b>Ca</b>	<b>Mg</b>	<b>Na</b>	<b>Cr</b>
<b>Alumina</b>							
no slag 1250 °C	1.3	97.9	0.0	0.4	0.1	0.2	N/A
F1 1250 °C	3.4	94.3	0.2	1.0	0.6	0.5	N/A
F3L1 1250 °C	1.8	97.0	0.0	1.1	0.0	0.0	N/A
F4L1 1250 °C	2.7	96.3	0.0	1.0	0.0	0.0	N/A
F1 1500 °C	2.5	95.9	0.3	0.6	0.4	0.4	N/A
F3L1 1500 °C	3.5	94.6	0.1	1.9	0.0	0.0	N/A
F4L1 1500 °C	2.8	95.5	0.1	0.9	0.5	0.3	N/A
<b>Silicon carbide</b>							
no slag 1250 °C	100.0	0.0	0.0	0.0	0.0	0.0	N/A
F1 1250 °C	99.9	0.0	0.1	0.0	0.0	0.0	N/A
F1 1500 °C	98.8	0.5	0.2	0.4	0.0	0.0	N/A
<b>Alumina-chromia</b>							
no slag 1250 °C	3.7	60.4	0.0	0.4	0.0	0.3	35.1
F1 1250 °C	14.8	24.2	5.4	4.9	7.8	4.1	38.8
F1 1500 °C	13.4	49.0	0.9	5.2	0.6	2.3	28.6

Figures 7.3a and 7.3b respectively, show F1 slag on alumina at 1250 and 1500 °C. Images for F3L1 and F4L1 on alumina are very similar. At both temperatures, the elemental maps and refractory compositions indicate very little slag penetrates the alumina. At 1250 °C there are no clearly distinguishable phases in all the slags with alumina refractory. By contrast, at 1500 °C all the slags caused alumina spalling, the F1 slag contains Al-Fe-Mg rich solids, while the F3L1 and F4L1 slags contain Al-Si-Ca rich solids. It can be seen in Table 7.6 that the Al content increases in F1 at 1250 °C, but not in F3L1 or F4L1. All three slags see an Al content increase at 1500 °C. The diffusion factors do not agree with these results when comparing the different slags at a given temperature. However, the diffusion factors do agree with the marked increase in Al diffusion with higher temperature. It is interesting to note that solids were present in the slags at 1500 °C and not at 1250 °C, which is the opposite of what was predicted by

FactSage calculations. This may be due to slow formation and dissolution of solids. Hence the metastable diffusion factor may be more appropriate than the non-metastable diffusion factor. Caution must be taken when comparing the atomic concentration of a single element in Table 7.6 as the concentrations are not only affected by diffusion from the refractory, but also by other changes in the slag composition. For example, the increased Al content of F1 from initial to 1500 °C is partially due to devolatilization of Na.

Figures 7.3c and 7.3d respectively, show F1 slag on silicon carbide at 1250 and 1500 °C. At both temperatures, the elemental maps and refractory compositions indicate very little slag penetrates the silicon carbide. Not enough slag adhered to silicon carbide at 1250 °C to obtain an accurate slag composition from SEM-EDX analysis. It was also noted that the slag is “bubbly”. This could be due to the reduction of iron to its metallic form. The elemental maps do show enrichment of iron at the slag-refractory interface. Iron enrichment is observed at the slag-silicon carbide interface at 1500 °C as well, yet the slag was not as “bubbly” and adhered well to the silicon carbide. Metallic iron may have been formed at 1500 °C, with the lower slag viscosity allowing the gas bubbles to escape easily. The observed iron accumulation at the interface supports the assumption of a highly reducing state near the silicon carbide, which was used for FactSage predictions when calculating the diffusion factors. At 1500 °C the slag is enriched in Si (29% increase). However about half of that increase can be attributed to the loss of Na which devolatilizes. This explains the slight enrichment in Al, Ca and Mg. About 98% of the iron is removed from the bulk slag, likely due to precipitation. The low diffusion factors (less than  $50 \text{ K}\cdot\text{Pa}^{-1}\cdot\text{s}^{-1}$ ) predicted for silicon carbide at 1500 °C compare well with the similar amount of diffusion and the low diffusion factors predicted for F1 with alumina at 1250 °C, as opposed to the higher amounts of diffusion observed and predicted (diffusion factors greater than  $100 \text{ K}\cdot\text{Pa}^{-1}\cdot\text{s}^{-1}$ ) for the slags on alumina at 1500 °C.

Figures 7.3e and 7.3f respectively, show F1 slag on alumina-chromia at 1250 and 1500 °C. The elemental maps indicate some slag penetrates the alumina-chromia at 1250 °C, while all the slag penetrated the alumina-chromia at 1500 °C. At 1250 °C, the slag seems

to have maintained the same composition as the original ash, and this compares well with the low metastable dissolution factor. Table 7.6 does not indicate a slag composition at 1500 °C since there was no slag left on top of the refractory. A Cr-Fe-Mg rich phase formed at the slag-refractory interface at both temperatures. Refractories with high Cr content are often used in gasifiers since the formation of spinel physically impedes slag penetration while also reducing Fe content in the liquid slag which increases its viscosity and further impedes penetration [24].

Overall, metastable diffusion factors correlate reasonably well with general trends observed experimentally. Based on the observations in this study, factors below  $50 \text{ K}\cdot\text{Pa}^{-1}\cdot\text{s}^{-1}$  correspond to little or no diffusion of refractory into slag. Factors above  $100 \text{ K}\cdot\text{Pa}^{-1}\cdot\text{s}^{-1}$  correspond to high diffusion of refractory in slag. The alumina-chromia refractory presents the worst slag corrosion resistance due to high levels of slag penetration. The alumina and silicon carbide refractories have very little slag penetration. Silicon carbide shows less spalling and diffusion into slag than alumina refractory. However, formation of metallic iron at the slag-silicon carbide interface may hinder slag flow by removing iron (II) oxide, which tends to reduce viscosity, from the bulk slag [6].

#### **7.4.4 CFD modeling**

The end goal of this three-part research program is to develop a gasifier CFD model which includes the effects of inorganic matter. Such a model requires many fuel properties as inputs, including the fuel properties used for screening, operation and design. The remaining properties are discussed here with characterization results made available as supplementary information. These results will be applied in the subsequent parts of the research program.

During volatilization, softening and/or swelling of coal, combined gases escaping will lead to the formation of porous char with possibly an overall cenospheric shape [32-35]. This will dictate whether inorganic matter in coal will coalesce or form separate entities. The structure of char can be predicted from the fuel form [33]. Coal petrography was

performed to determine the abundance of maceral types of the fuels in this study and the results are presented in Table S1 (supplementary data). Fuel particle size is important to predict ash particle formation and gas-particle transport. If the Weber number is applied to model particle sticking, then particle size is needed for this as well. Table S2 (supplementary data) presents the size distributions of the fuels and limestone.

The form, particle size and associations of inorganic matter in the fuel also play a key role in predicting ash particle formation [32-35]. The inorganic matter form indicates whether it is crystalline or amorphous, and the crystal structure/composition if applicable. At room temperature, the form is best characterized by XRD analysis since the alternative, FactSage modeling, does not account for the extremely slow kinetics. XRD results for each fuel and the limestone selected are given in Table S3 (supplementary data). At elevated temperatures, HT-XRD can be applied but its accuracy is compromised as the liquid content rises. In this case, FactSage predictions may be more appropriate. FactSage predictions for F1, F3L1 and F4L1 are given in Tables S4, S5 and S6 (supplementary data), respectively. The associations indicate whether the inorganic matter is a particle included in a coal particle, a particle excluded from coal particles, or organically bound to the coal matter. As was the case for the fuel particle sizes, the inorganic matter particle sizes will also affect gas-particle transport and sticking. CCSEM was performed with each fuel to determine the size and association of their inorganic features. Tables S7, S8, S9 and S10 (supplementary data) contain data on inorganic features detected by CCSEM in F1, F3, F4 and L1, respectively. This data includes the area, average diameter, composition and association for each feature. If the inorganic feature was detected within a coal particle, the indicated association is *included*. Otherwise, the indicated association is *excluded*.

Slag interfacial tension can be used when predicting particle sticking since it is present in the Weber number. It may also be used when modeling slag-refractory interaction since wetting and penetration of slag onto or into the refractory are functions of the interfacial energy. The reader is referred elsewhere [36,37] for thorough reviews of theory, measurement techniques, data and models relating to slag interfacial tension. For a

particle sticking model, Yong [10] applied an equation presented by Mills and Rhine [36] for slag interfacial tension with gas. The equation relates the interfacial tension to temperature and slag composition. A contact angle is typically used to describe interfacial tension for slag on a solid or liquid. Some contact angles for various slags on different substrates is available in the Slag Atlas [37]. To model slag deposition in a gasifier, Ni *et al.* used fitted equations for contact angles of slag on a gasifier wall as a function of temperature [13]. It was noted that for two coal ashes, the contact angle is relatively constant at about 70° when the substrate is above 500 °C. In another gasifier slagging model, Ni *et al.* used a constant contact angle of 67° for slag on a gasifier wall [38]. Contact angle measurements may be required for more accurate predictions with specific slag-substrate combinations and conditions.

The density of inorganic matter is involved in all the inorganic behaviour phenomena. The density of solid inorganic matter can be obtained from literature or FactSage. The reader is again referred elsewhere [36,37] for thorough reviews of theory, measurement techniques, data and models relating to liquid slag density. The density of the liquid slag can be calculated from the partial molar volumes of its components. Partial molar volumes for various components are available from [10] and [39].

Many heat transfer sub-models must be applied when modeling gasification. Temperature is very important as it will affect many parameters in the other sub-models such as reaction rates and viscosity predictions. Fletcher *et al.* [40], Lipinski and Steinfeld [41] and Watanabe and Otaka [42] provide equations for heat transfer between the gas phase and the particles. They considered convective heat transfer, latent heat transfer associated with mass transfer, heat from the gasification reactions, and radiative heat transfer between the particles and the gasifier walls. The reader is also referred elsewhere for thorough reviews of heat transfer theory, measurement techniques, data and models relating to ash and slag [37,43-45]. Thermal conductivity of slag is a function of temperature and composition. Wang *et al.* [46] provide a correlation for thermal conductivity of a running slag layer. However, for modeling slag behaviour in reactors, it is common to express liquid slag's thermal conductivity as a function of its thermal

diffusivity, heat capacity and density [44]. Seggiani [20], Ni *et al.* [38] and Yong [10] used this approach with a thermal diffusivity of  $4.5 \times 10^{-7} \text{ m}^2/\text{s}$ . Ni *et al.* state that the thermal conductivity is lower for solid slag than it is for liquid slag, citing a value for solid slag of  $0.6 \text{ W/m}^\circ\text{C}$  from [43]. As for emissivity, Mills and Rhine [44] obtained a constant emissivity of 0.83 for a slag in the temperature range 1343-2073 °C. This value was applied by Seggiani [20] and Ni *et al.* [38]. Mueller *et al.* applied a constant emissivity of 0.75 [14]. Wang *et al.* [46] applied a constant emissivity of 0.4. Measurements for the conditions of interest may be required. Determining slag heat capacity can be very difficult due to multiple phases and phase transitions. Inorganic solids may contain distinct crystals which are not necessarily the equilibrium phases. In this case, the crystals present must be known and the overall heat capacity can be a weighted average of the individual crystal heat capacities. However, if the ash is fully molten or has been fully molten and slowly cooled, then the heat capacity can be obtained from an equation for the equilibrium heat capacity. One method to estimate a slag's equilibrium heat capacity is to use FactSage. Alternatively, Mills and Rhine [44] suggest using the Kopp-Neumann rule to estimate the heat capacity. Ni *et al.* [38] and Yong [10] applied this approach.

## 7.5 Conclusions

Thorough characterization of four coals (F1-F4) and a limestone (L1) was performed to evaluate properties for fuel screening, gasifier operation, gasifier design and gasifier CFD modeling. Focus was on the properties that relate to the fate of inorganic matter in entrained-flow slagging gasifiers. Fuel composition, inorganic matter composition and ash melting temperatures were applied for screening. Based on these properties it was determined that F2 is not a good fuel candidate and therefore would not be evaluated in subsequent testing. Slag viscosity was modeled and measured for the remaining fuels to determine optimal fluxing ratios for gasifier operation. Assuming a slag tap temperature of 1400 °C and a desired viscosity below 10 Pa·s at that temperature, F1 does not require

fluxing with L1. The optimal blend of F3 and L1 (F3L1) contains 0.042 kg L1 per kg F3. The optimal blend of F4 and L1 (F4L1) contains 0.055 kg L1 per kg F3. Slag-refractory reactivity was modeled and measured for the fuels or optimal fuel blends to assess the suitability of three refractory materials for gasifier design. Alumina, silicon carbide and alumina-chromia (30 wt.% chromia) were evaluated as potential refractory materials. The alumina-chromia refractory presents the worst slag corrosion resistance. Alumina is preferred over silicon carbide despite presenting slightly more spalling and dissolution in slag due to metallic Fe accumulation at the interface between silicon carbide and slag. Finally, the fuel forms and particle sizes, as well as the inorganic matter forms, particle sizes, associations, interfacial tension, density, heat capacity, heat conductivity and emissivity were discussed in terms of their application to CFD modeling. Experimental characterization or FactSage modeling is desired for some of these properties and the results are supplied as supplementary data for this paper.

## **7.6 Acknowledgements**

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## **Chapter 8. Fate of inorganic matter in entrained-flow slagging gasifiers: Pilot plant testing**

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## 8.1 Abstract

This study is the second of a three-part research program that involves fuel characterization, testing in a pilot-scale gasifier, and computational fluid dynamics (CFD) modeling for entrained-flow slagging gasification. Focus of this paper is on the behaviour of inorganic fuel components since the end goal is to develop a CFD model which includes inorganic matter transformations. Using CanmetENERGY's 1 MW<sub>th</sub> gasifier, five gasification tests have been completed with three different coal fuels and a limestone as fluxant. The refractory liner from the upper section of the gasifier was removed for analysis after each test. The liners were made of sintered alumina or castable alumina-chromia refractory. Carbon conversions and cold gas efficiencies for the gasification tests ranged within 97.0-99.6% and 33.7-61.9%, respectively. Solid samples from the refractory liners, in-situ gas sampling probe sheaths and impingers, the slag tap, the slag pot, quench discharge water and scrubber water were collected and characterized. Char and fly ash samples indicate that most of the inorganic matter melted and formed spheres. Devolatilization and interaction with the gasifier refractory affected the composition of slag collected from the gasifier. Signs of refractory spalling/erosion were detected. The slag layers formed on the alumina liners are smooth with some rivulets and spotting. The slag layers formed on the alumina-chromia liners are rough and bubbly. Slag penetration fractions were determined for all liners.

*Keywords:* Gasification, Inorganic, Mineral, Ash, Slag, CFD model

## 8.2 Introduction

Gasification is a flexible technology which is applied in industry for electricity generation, hydrogen production, steam raising and liquid fuels production [1]. Furthermore, it can utilize one or more feedstocks such as coal, biomass, municipal waste and petroleum coke. This versatility, in addition to being adaptable to various emissions control technologies, including carbon capture, renders it an attractive option for years to come [2]. Most of the successful high throughput coal gasifiers developed in the past 60 years are of the entrained-flow slagging type [1]. This study is the second part of a three-part research program which involves fuel characterization, testing in a 1 MW<sub>th</sub> gasifier, and computational fluid dynamics (CFD) modeling for entrained-flow slagging gasification. The focus of this program is on the behaviour of inorganic fuel components in the gasifier as this is still ill-understood even though it can be the determining factor in designing and operating entrained-flow gasifiers [3]. Results from fuel characterization have been published elsewhere [4].

The motivation for pilot-scale testing is to provide data for a comprehensive CFD model which includes ash particle formation, gas-particle transport, particle sticking, slag flow and slag-refractory interaction. Multiple studies have been conducted with pilot scale or larger gasifiers [3,5-8]. Emphasis is often on the syngas composition and fuel conversion. Limited experimental data is provided on the ash and slag produced. The current study investigated inorganic matter phenomena throughout the gasification process by systematically collecting samples which represent each step of ash and slag transformations. These samples were characterized by various techniques to obtain data for CFD model validation. The pilot testing program was based on results from fuel and limestone characterization which considered screening of potential fuels, gasifier operation, gasifier design and CFD modeling inputs [4]. Of the four coals that were characterized, one was eliminated at the screening step due its high ash content combined with its requirement of a fluxing agent. The three fuels remaining were coded F1, F3 and

F4. F1 fuel is a lignite coal from Saskatchewan, Canada. F3 fuel is a beneficiated sub-bituminous coal from Alberta, Canada. F4 fuel is another sub-bituminous coal from Alberta, Canada. Within the fuel characterization study, it was determined that F1 does not require fluxing for proper gasifier operation, while optimal fluxing ratios with L1, a limestone from a Canadian power company, were determined for F3 and F4. These optimal blends for F3 and F4 fuels were dubbed F3L1 and F4L1, respectively. The intention was to use F1, F3L1 and F4L1 for the pilot plant tests. However, the available quantity of L1 was insufficient. L2, a limestone from Nova Scotia, Canada, was used instead of L1. Due to their similar compositions, the same fuel-to-limestone mass ratios were used for F3L2 and F4L2 as for F3L1 and F4L1, respectively. Inorganic matter composition, particle size distribution, X-ray diffraction analysis and computer-controlled scanning electron microscopy analysis for L2 limestone are available in supplementary data Tables S1-S4, respectively. Experimental methods for these results are as described by Duchesne *et al.* [4].

In total, five pilot-scale gasifier tests were performed: tests T1-T3 were with F1 fuel, test T4 was with F3L2 fuel and test T5 was with F4L2 fuel. The operating conditions and gasifier design used are described and justified based on fuel characterization and operational experience. Mass balances on ash and carbon were performed, and subsequently carbon conversion and cold gas efficiency were determined for each test. Solid samples were collected from the refractory liner, the gas probe sheath and impingers, the slag tap, the slag pot, quench discharge water and scrubber water. In addition to validation data for CFD models, characterization of these samples provides insight on often overlooked features of inorganic matter transformations in entrained-flow gasifiers.

## 8.3 Experimental

### 8.3.1 Gasifier system

The CanmetENERGY pressurized entrained-flow slagging gasifier was configured for single-stage slurry-fed gasification as shown in Figure 8.1. Water, coal and limestone (when required) were combined in the slurry tank in sufficient quantity to complete each gasification test. The slurry components were vigorously mixed using both a helical mixer that sweeps the full volume of the tank and a slurry circulation pump with its suction at the bottom dish of the tank with a flow rate sufficient to provide a minimum of eight tank changeovers per hour. Slurry samples were taken before and after each test to ensure that the moisture content and slurry viscosity remained constant throughout the test. During gasification tests, the slurry was pressurized and metered into the gasifier using a progressing cavity pump. The slurry flow rate was determined by loss-in-weight from the slurry tank based on the feedback from load cells with a resolution of +/- 0.5 kg. Oxygen was supplied from a cryogenic oxygen tank equipped with a vapourizer. The oxygen flow rate was measured using thermal mass flow meters and controlled by plug valves. Automation and data collection for the pilot gasifier facility is accomplished using an ABB Freelance 2000 distributed control system. No steam or other moderators were injected during the gasification tests.

Slurry and oxygen were premixed within the down-fired gasifier burner in a Delavan Air Swirl nozzle providing a finely atomized hollow cone spray and subsequent flame in the gasifier reactor which was operated at 1.5 MPa(gauge). Details of the slurry spray characteristics under these conditions including spray angle, droplet size distribution, and axial mass flux at various elevations and radial positions are provided by Daviault et al. [9].

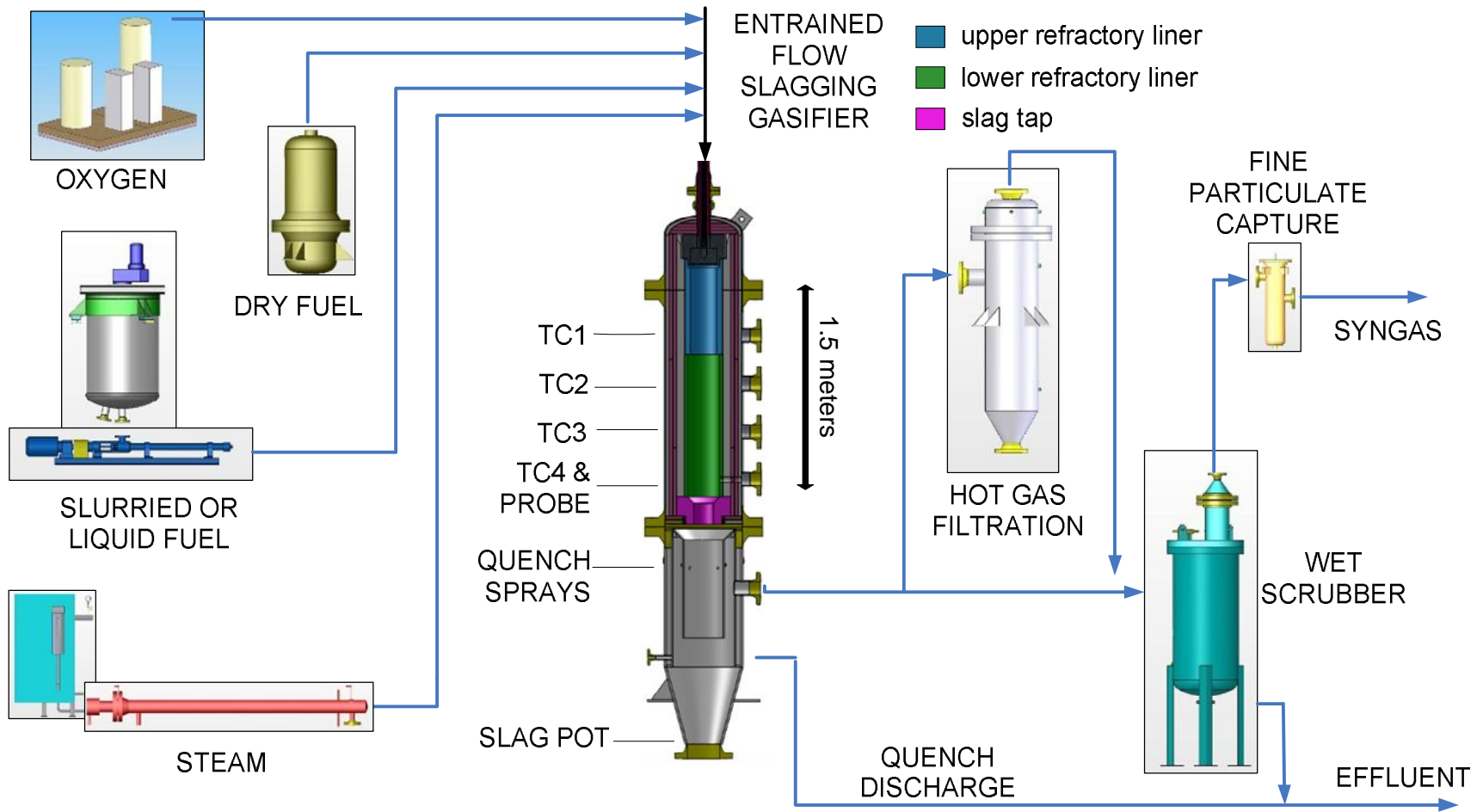
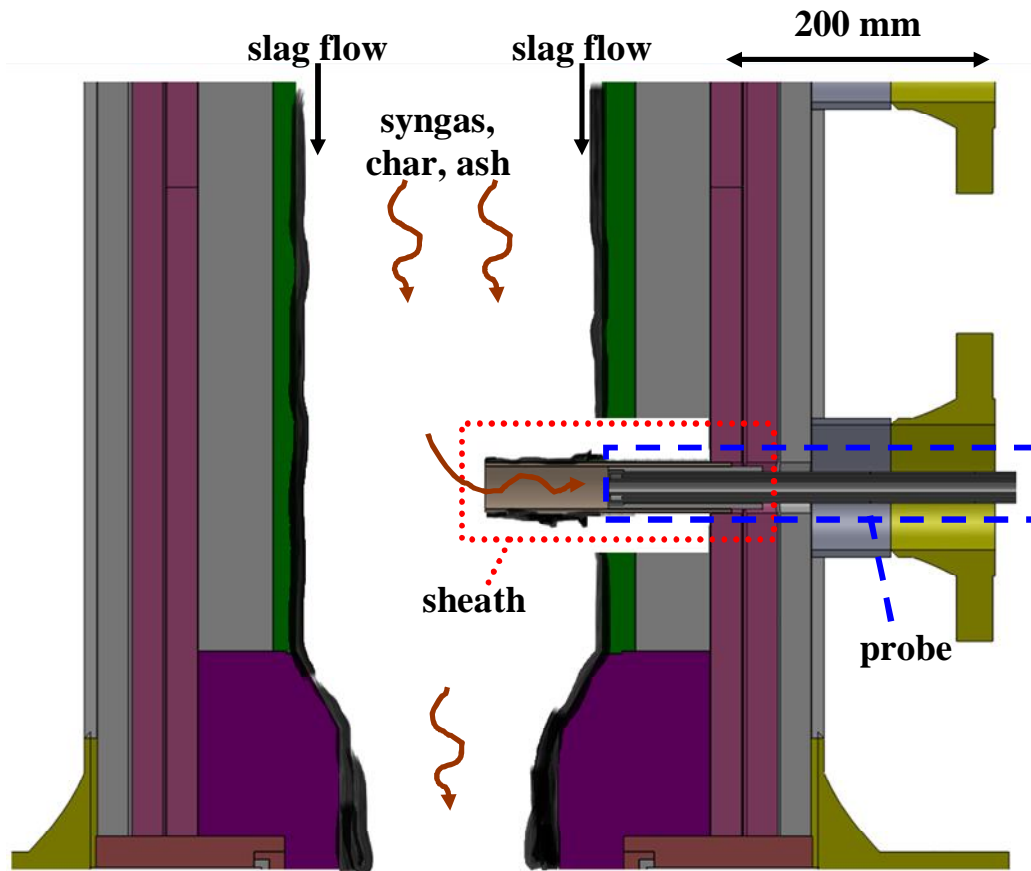


Figure 8.1: Schematic diagram of CanmetENERGY's pressurized entrained-flow gasification system.

The insulation and refractory system from the gasifier shell to the gasifier hot face is made up of the following layers: alumina paper insulation, alumina board insulation, alumina bubble castable refractory and finally the hot face. The hot face is made up of three sections. The upper section shown in blue (upper refractory liner) in Figure 8.1 is removed after each test. In tests T1 and T2 this section was a high density sintered alumina liner with an inner diameter of 250 mm, thickness of 9.5 mm and length of 730 mm. In tests T3, T4, and T5 this section was a 30% Cr<sub>2</sub>O<sub>3</sub> balance Al<sub>2</sub>O<sub>3</sub> cast refractory with an inner diameter of ~200 mm, thickness of 40 mm and length of 730 mm. The two lower sections shown in green (lower refractory liner) and pink (the slag tap) were chrome/alumina cast refractory. The lower refractory liner had an inner diameter of ~250 mm while the slag tap reduced down to ~200 mm diameter. The temperature profile within the gasifier was monitored using type B thermocouples with the tip of the thermocouples protruding past the hot face and into the reaction chamber by ~10 mm. The locations of the thermocouples are indicated on Figure 8.1 as TC1 through TC4. The burner shutdown sequence included starting the injection of 30 to 60 kg/hr of water immediately prior to diverting slurry from the gasifier and isolating the oxygen flow to the gasifier. The water flow into the gasifier was maintained until the gasification chamber temperature was less than 1000 °C. This procedure was followed in order to rapidly quench the upper refractory liner and the probe sheath (described below). After each gasification test, the slag was bored out of the slag tap for analysis using a 200 mm diameter diamond tipped coring bit.

A nitrogen-cooled gas sampling probe was inserted into the gasifier at the same elevation as TC4 for each of the tests (Figure 8.2). Insulation was placed on the probe and then covered by a sheath. The sheath (length 180 mm, inner diameter 35 mm, thickness 3.2 mm) protruded past the hot face and into the reaction chamber by ~75 mm. Finite element heat transfer analysis performed during experimental design has shown that the outer surface temperature of the sheath is within a few degrees Celsius of the prevailing syngas temperature and TC4 likely represents the temperature of the sheath quite well. The sheath and probe were retracted from the gasification chamber using a remotely operated servo motor immediately upon burner shut down in order to rapidly freeze the slag for subsequent analysis. During the tests, syngas was extracted through the probe

and rapidly cooled in a tube-in-tube heat exchanger to provide the syngas composition near the probe sheath. The cooled syngas was passed through a 10 micron filter to collect fine particulates, a series of water-cooled impingers, and a nitrogen-swept nafion drying membrane prior to gas analysis. Gas analysis was performed via gas chromatography for each of the tests. On-line infrared carbon monoxide and carbon dioxide analyzers were used to validate syngas compositions generated by the gas chromatograph.



**Figure 8.2: Schematic diagram of the gas probe inserted into the gasifier.**

Coarse slag that flowed out of the gasifier was quenched with water and collected in a vessel at the bottom of the quench vessel. The coarse slag collection vessel is identified as the slag pot in Figure 8.1. Quench water along with fine ash and char flowed out of the quench vessel into a bag filter (10 micron) in which the particulate matter was captured for analysis. Particulate matter and water from the wet scrubber passed into a similar bag filter to capture solids for analysis. After passing through the water scrubber,

the syngas was passed through a bag filter to ensure that no particulate matter escaped from the system. Insufficient particulate matter was collected from this bag filter to perform analysis and hence it is not discussed further in this paper.

### **8.3.2 Surface roughness measurements**

The setup used for surface roughness measurements is shown in Figure 8.3. The upper refractory liners were first sectioned into the top and bottom halves, and each half was then cut into three 120° sections. These sections were placed on the rotating table shown in Figure 8.3, with shims and a bracket to keep them level. A Micro-Epsilon scanCONTROL LLT2800-100 laser is mounted on a two-axis rail system including a vertical stage and a horizontal stage. The laser is connected to a scanCONTROL controller. The controller communicates with a personal computer via a FireWire (IEEE 1394) link. The Export Profiles program of scanCONTROL Configuration Tools 1.1 software was used to collect and export surface profiles (i.e., a measurement series of horizontal distances between the laser and test piece surface) which were manipulated using Microsoft Excel. For all profiles collected in this study, the laser was placed at a maximum horizontal distance of 230 mm from the test piece surface to be within the laser's standard measuring range. The laser filter settings were set to have a large range with interpolation and re-sampling activated. Surface roughness is determined from a surface profile (1024 measurements over 80 mm) using the arithmetic average of absolute deviations from the mean distance [10].

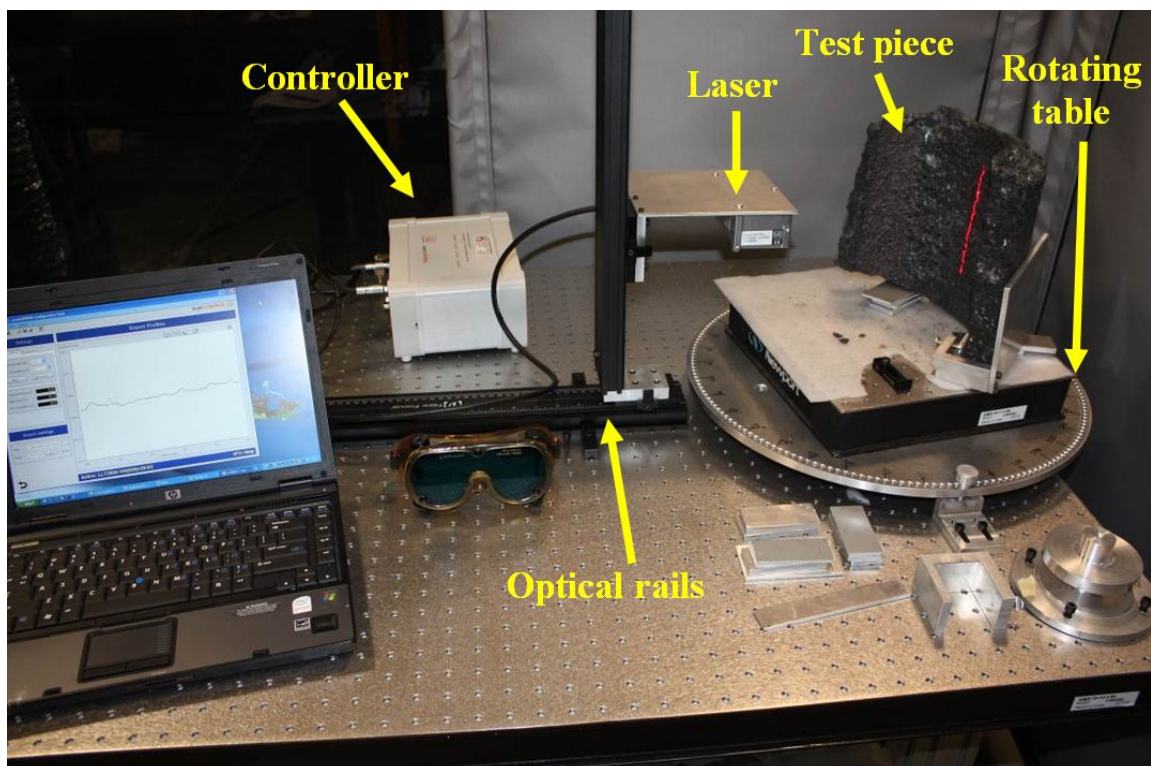


Figure 8.3: Surface roughness measurement system.

### 8.3.3 Surface area and density

Nitrogen adsorption isotherms for scrubber and probe impinger samples were obtained at  $-196\text{ }^{\circ}\text{C}$  on a Micromeritics Tristar II 3020 porosimetry system. The Brunauer Emmett Taylor (BET) model was used to determine the surface area. The Barret-Joyner-Halenda (BJH) model was used to determine the pore volume. Skeletal density of scrubber and crushed slag tap samples was determined using the Micromeritics AccuPyc II 1340 helium pycnometry system. Envelope density of slag tap samples was determined by placing  $\sim 7\text{ g}$  of the sample in a  $25\text{ cm}^3$  graded cylinder and pipetting  $10\text{ cm}^3$  of water into the cylinder. Measurements were duplicated and the average taken. Envelope density of scrubber samples was calculated from skeletal density and pore volume. Scrubber and slag tap porosities were calculated from skeletal and envelope densities.

### **8.3.4 Other methods**

ASTM D7582 was applied for determination of ash content in solid samples. ASTM D5373 was applied for determination of carbon content in solid samples, except for probe samples since the amount collected was insufficient for this procedure. Alternatively, ASTM D5291 was applied for determination of carbon content in samples from the probe. ASTM D4326 was applied to determine major and minor oxides content via X-ray fluorescence (XRF).

Quantitative X-ray diffraction (XRD) analysis of the samples was performed over the angular range of 7 to 80° (2 $\theta$ ). Each sample was mixed with an internal standard. Integrated intensities for strong diffracting peaks of each phase were obtained using JADE version 7.0 XRD processing software. Concentrations of the chemicals identified in each sample were calculated using the reference intensity ratio (RIR) method.

Scanning electron microscopy with energy dispersive X-ray spectroscopy (SEM-EDX) analysis was completed for cross-sections of the slag-covered refractory and solids from the probe impingers and scrubber water. Samples were mounted with carbon tape on an aluminum holder. A Hitachi 3400N VP-SEM with Oxford Instruments Si(Li) Pentafet Plus X-Max 20 mm<sup>2</sup> detector was used. X-ray acquisition was done with Oxford AZtec Point and ID software. Quantification was done with the base Oxford factors.

<sup>57</sup>Fe Mössbauer spectroscopy measurements were conducted at room temperature using a standard Mössbauer spectrometer operating in a sine mode and with a <sup>57</sup>Co(Rh) gamma-ray source. The spectrometer was calibrated with a 6.35- $\mu$ m-thick  $\alpha$ -Fe foil.

## 8.4 Results and Discussion

### 8.4.1 *Description of pilot plant tests*

Five gasification tests were conducted. Details for each are provided in Table 8.1. The tests lasted 2-4 h. Injection rates, discharge rates, temperatures and dry syngas compositions presented in Table 8.1 are averages for the final hour of operation. These are representative of the quasi-steady state operating conditions during each test in terms of feed rates and syngas composition, and are most representative of the conditions prior to system quenching at the end of each test. The gasifier design allowed removal of the upper refractory liner after each test. An alumina liner was used for tests T1 and T2. An alumina-chromia (30 wt.% chromia) liner was used for tests T3-T5.

For each test, solid samples were collected from gas probe impingers, the slag tap, the slag pot, the quench water discharge and the scrubber. The ash content, carbon content and mass of the dried solids are given in Table 8.2. There was insufficient sample for analysis from the scrubber for test T2, and from the probe for tests T1 and T5. Using the results in Table 8.2 in conjunction with the total slurry mass consumed and composition, ash and carbon mass balances were performed for each gasification test. Results of the ash and carbon mass balances are presented in Tables 8.3 and 8.4, respectively. Note that the mass of ash accumulated in the gasifier was calculated by difference. It was assumed that accumulated ash in the gasifier has the same carbon content as ash at the slag tap. The carbon in the syngas was calculated by difference and was used to calculate the total mass of syngas produced. The carbon conversion and cold gas efficiency were calculated for each test and are given in Table 8.1.

**Table 8.1: Conditions for pilot-scale gasifier tests**

<b>Test name</b>	<b>T1</b>	<b>T2</b>	<b>T3</b>	<b>T4</b>	<b>T5</b>
<b>Fuel</b>	F1	F1	F1	F3	F4
<b>Upper refractory liner material</b>	alumina	alumina	alumina- chromia	alumina- chromia	alumina- chromia
<b>Amount L2 per amount fuel (kg/kg)</b>	0.000	0.000	0.000	0.042	0.099
<b>Slurry moisture (wt.%)</b>	58.0	56.0	59.5	54.6	52.1
<b>Slurry used (kg)</b>	200	156	197	257	158
<b>Test duration (h)</b>	2.46	2.62	2.85	3.36	2.05
<b>Slurry injection rate (kg/h)</b>	82.9	66.0	75.6	81.5	80.6
<b>Oxygen injection rate (kg/h)</b>	43.8	33.9	37.1	48.3	43.1
<b>Quench water injection rate (kg/h)</b>	302	214	172	350	327
<b>Quench water discharge rate (kg/h)</b>	367	291	241	389	369
<b>TC1 temperature (°C)</b>	1466	1593	1609	1554	1582
<b>TC2 temperature (°C)</b>	1413	1562	1441	1476	1501
<b>TC3 temperature (°C)</b>	1381	1422	1339	1440	1452
<b>TC4 temperature (°C)</b>	1285	1254	1253	1385	1371
<b>Dry syngas composition</b>					
<b>H<sub>2</sub> (vol.%)</b>	25.1	21.7	24.0	29.6	23.1
<b>CO (vol.%)</b>	28.0	20.9	24.5	41.9	25.0
<b>CO<sub>2</sub> (vol.%)</b>	46.1	56.3	50.6	28.0	50.7
<b>Carbon conversion<sup>a</sup> (%)</b>	99.5	98.7	97.0	99.6	97.8
<b>Cold gas efficiency<sup>b</sup> (%)</b>	44.1	33.7	38.8	61.9	39.3

<sup>a</sup> Based on carbon content of solids.

<sup>b</sup> Based on higher heating values.

A very small amount of solids was entrained with the syngas to the scrubber. Most of the solids were collected from the slag tap and slag pot. However, the majority of the solids in tests T1 and T2 accumulated in the gasifier. Carbon conversion was 97.0% or greater for every test. Most of the unconverted carbon was found in the slag pot. Cold gas efficiencies varied from 33.7 to 61.9%. It should be noted that optimization of conversion and efficiency was not the primary objective of the gasifier tests, but rather steady gasification conditions for analysis of the inorganic matter were desired.

**Table 8.2: Ash content, carbon content and mass of dried collected solids**

<b>Test</b>	<b>Source</b>	<b>Ash (wt.%)</b>	<b>Carbon (wt.%)</b>	<b>Mass (g)</b>
T1	scrubber	20.0	71.9	19
T2	scrubber	3.2	85.2	7
T4	scrubber	77.0	20.0	95
T5	scrubber	18.1	73.8	9
T2	probe	N/A	1.7	< 1
T3	probe	N/A	24.7	< 1
T4	probe	N/A	31.1	< 1
T1	quench discharge	91.2	9.0	146
T2	quench discharge	33.0	51.4	990
T3	quench discharge	53.5	37.6	447
T4	quench discharge	81.3	19.2	338
T5	quench discharge	78.1	18.8	165
T1	slag tap	98.1	1.7	3288
T2	slag tap	100.0	0.1	3363
T3	slag tap	100.0	0.0	2901
T4	slag tap	100.0	0.0	5775
T5	slag tap	100.0	0.0	2264
T1	slag pot	96.3	3.5	2994
T2	slag pot	66.1	26.2	113
T3	slag pot	83.8	14.1	9282
T4	slag pot	96.3	3.3	8144
T5	slag pot	94.7	4.4	18940
T1	slag pot (chunks)	100.0	0.0	N/A
T2	slag pot (chunks)	98.3	1.3	N/A
T3	slag pot (chunks)	99.1	0.7	N/A
T4	slag pot (chunks)	100.0	0.0	N/A
T5	slag pot (chunks)	100.0	0.0	N/A
T1	slag pot (sheets)	100.0	0.0	N/A
T5	slag pot (sheets)	99.8	0.2	N/A

**Table 8.3: Ash mass balance**

	<b>T1</b>	<b>T2</b>	<b>T3</b>	<b>T4</b>	<b>T5</b>
<b>Total in (g)</b>	12372	10134	11778	14816	22009
<b>Slag tap (g)</b>	3241	3364	2901	5775	2263
<b>Slag pot (g)</b>	2884	75	7782	7847	17944
<b>Quench discharge (g)</b>	133	327	239	275	129
<b>Scrubber (g)</b>	4	0	0	73	2
<b>Accumulated in gasifier (g)</b>	6110	6369	856	845	1672

**Table 8.4: Carbon mass balance**

	<b>T1</b>	<b>T2</b>	<b>T3</b>	<b>T4</b>	<b>T5</b>
<b>Total in (g)</b>	51796	42427	49308	78961	38970
<b>Slag tap (g)</b>	44	2	0	0	0
<b>Slag pot (g)</b>	105	30	1306	268	834
<b>Quench discharge (g)</b>	13	509	168	65	31
<b>Scrubber (g)</b>	13	6	0	19	7
<b>Accumulated in gasifier (g)</b>	104	6	0	0	0
<b>Syngas (g)</b>	51516	41874	47833	78608	38099

### **8.4.2 Char and fly ash**

Solid samples collected from the gas probe impingers are representative of char and fly ash ~150 mm above the slag tap. Samples collected from the scrubber are representative of char and fly ash which were entrained out of the reactor by syngas. Ash content, carbon content, as well as mass of samples collected from the scrubber and probe are given in Table 8.2. The probe samples contain 2-31 wt.% carbon. Scrubber samples contain 20-85 wt.% carbon. Ash compositions of the solid samples, on a major and minor oxides basis, are provided in Table 8.5. Fuel ash compositions are included in this table for comparison. Quantitative XRD analyses of the solid samples are provided in Table 8.6. Masses of probe samples were insufficient for the characterization in Tables 8.5 and 8.6. Compared to the fuel ashes, scrubber samples from tests T1 and T2 are enriched in

**Table 8.5: Major and minor oxides in ash of collected solid samples (part 1 of 2)**

<b>Test</b>	<b>Source</b>	<b>SiO<sub>2</sub></b> <b>(wt.%)</b>	<b>Al<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>	<b>Fe<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>	<b>TiO<sub>2</sub></b> <b>(wt.%)</b>	<b>P<sub>2</sub>O<sub>5</sub></b> <b>(wt.%)</b>	<b>CaO</b> <b>(wt.%)</b>	<b>MgO</b> <b>(wt.%)</b>	<b>SO<sub>3</sub></b> <b>(wt.%)</b>	<b>Na<sub>2</sub>O</b> <b>(wt.%)</b>	<b>K<sub>2</sub>O</b> <b>(wt.%)</b>	<b>Cr<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>
T1	scrubber	50.3	22.1	6.7	0.7	0.3	5.5	0.6	4.4	5.1	1.7	1.1
T2	scrubber	49.8	22.0	14.4	0.0	0.0	7.9	0.0	0.0	0.0	0.7	2.9
T4	scrubber	45.2	19.3	4.9	0.7	0.8	23.6	1.7	0.9	0.6	1.1	0.1
T5	scrubber	43.4	20.6	7.6	1.0	0.8	15.7	0.7	7.0	0.0	1.2	1.2
T1	quench discharge	52.8	22.3	4.3	0.8	0.2	8.0	2.0	0.5	6.4	1.6	0.2
T2	quench discharge	51.8	20.7	5.5	0.7	0.2	9.6	1.4	5.0	2.9	1.2	0.1
T3	quench discharge	48.4	25.7	4.9	0.7	0.2	9.6	2.3	1.7	4.4	1.2	0.2
T4	quench discharge	48.4	18.2	5.0	0.6	0.8	22.1	1.5	0.6	0.7	1.5	0.1
T5	quench discharge	54.1	17.0	4.4	0.5	0.1	18.3	1.0	0.5	1.3	2.1	0.1
T1	slag tap	51.4	21.5	5.5	0.8	0.2	11.0	2.4	0.0	5.0	1.1	0.1
T2	slag tap	49.5	24.3	5.5	0.8	0.2	9.8	2.4	0.0	5.0	1.1	0.5
T3	slag tap	48.5	22.4	5.8	0.8	0.2	10.4	2.7	0.0	5.1	1.1	2.1
T4	slag tap	46.8	19.2	5.8	0.5	0.4	21.5	1.8	0.0	0.9	1.0	1.5
T5	slag tap	49.0	20.4	5.0	0.5	0.1	18.1	1.4	0.0	1.5	1.2	2.4

**Table 8.5: Major and minor oxides in ash of collected solid samples (part 2 of 2)**

<b>Test</b>	<b>Source</b>	<b>SiO<sub>2</sub></b> <b>(wt.%)</b>	<b>Al<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>	<b>Fe<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>	<b>TiO<sub>2</sub></b> <b>(wt.%)</b>	<b>P<sub>2</sub>O<sub>5</sub></b> <b>(wt.%)</b>	<b>CaO</b> <b>(wt.%)</b>	<b>MgO</b> <b>(wt.%)</b>	<b>SO<sub>3</sub></b> <b>(wt.%)</b>	<b>Na<sub>2</sub>O</b> <b>(wt.%)</b>	<b>K<sub>2</sub>O</b> <b>(wt.%)</b>	<b>Cr<sub>2</sub>O<sub>3</sub></b> <b>(wt.%)</b>
T1	slag pot	51.9	20.5	5.6	0.8	0.2	11.0	2.6	0.0	5.1	1.2	0.1
T2	slag pot	52.0	21.0	5.7	0.8	0.2	10.3	2.3	0.8	4.5	1.2	0.3
T3	slag pot	48.4	23.7	5.4	0.7	0.2	10.5	2.3	0.2	4.8	1.1	1.8
T4	slag pot	45.2	22.0	5.6	0.5	0.4	20.2	1.7	0.2	1.0	1.1	1.6
T5	slag pot	53.6	16.6	4.9	0.5	0.1	19.6	1.2	0.0	1.3	1.5	0.4
T1	slag pot (chunks)	51.9	20.8	5.5	0.8	0.2	10.8	2.7	0.0	5.2	1.1	0.1
T2	slag pot (chunks)	51.8	21.5	5.8	0.8	0.2	10.2	2.5	0.0	5.0	1.1	0.2
T3	slag pot (chunks)	50.5	21.6	5.4	0.8	0.2	10.9	2.6	0.0	5.3	1.1	0.6
T4	slag pot (chunks)	48.7	19.2	6.0	0.6	0.3	17.8	2.1	0.0	2.3	1.1	1.3
T5	slag pot (chunks)	54.5	15.9	4.8	0.5	0.1	19.8	1.2	0.0	1.3	1.5	0.2
T1	slag pot (sheets)	51.5	20.7	5.5	0.8	0.2	11.2	2.5	0.0	5.3	1.2	0.1
T5	slag pot (sheets)	21.1	51.2	3.0	0.2	0.1	7.9	1.0	0.0	0.7	0.4	14.2
T1	slag pot (filaments)	52.5	20.6	5.4	0.8	0.2	11.2	2.2	0.0	5.0	1.2	0.0
T5	slag pot (filaments)	54.8	15.7	4.6	0.5	0.0	19.9	1.1	0.0	1.3	1.5	0.1
T1-T3	F1 fuel	38.0	16.3	6.3	0.7	0.4	13.6	2.7	14.3	6.0	0.3	0.0
T4	F3:L2 fuel	45.6	15.2	5.6	0.5	0.5	25.7	1.6	3.8	0.3	1.1	0.0
T5	F4:l2 fuel	48.4	16.6	3.9	0.6	0.0	21.5	1.4	4.1	1.9	1.5	0.0

**Table 8.6: Quantitative XRD analyses for solid samples (values reported as wt.%)**

	scrubber				slag tap					slag pot					slag pot, filaments	
	T1	T2	T4	T5	T1	T2	T3	T4	T5	T1	T2	T3	T4	T5	T1	T5
CaAl <sub>2</sub> Si <sub>2</sub> O <sub>8</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	2.7	0.0	0.0	0.0	0.0
(Ca,Na)(Al,Si) <sub>2</sub> Si <sub>2</sub> O <sub>8</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.8	0.0	0.0	0.0	0.0	0.0
(Mg,Fe,Ti,Al)(Ca,Na,Mg, (Si,Al) <sub>2</sub> O <sub>6</sub>	0.0	0.0	0.0	0.0	6.1	16.7	8.4	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca(CO <sub>3</sub> )	2.3	0.9	0.0	1.6	0.6	0.0	0.0	0.0	0.0	0.0	0.0	0.0	8.1	0.2	0.0	0.0
Ca(SO <sub>4</sub> )(H <sub>2</sub> O) <sub>0.6</sub>	5.2	0.6	0.0	4.3	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
FeCr <sub>2</sub> O <sub>4</sub>	0.0	0.0	0.0	0.0	0.0	1.9	1.8	1.7	3.2	0.0	0.0	0.0	0.0	0.0	0.0	0.0
CrO <sub>2</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.9	0.0	0.0	0.0	0.0
Al <sub>2</sub> O <sub>3</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	5.3	8.0	0.0	0.0	0.0
Cr <sub>2</sub> O <sub>3</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	4.0	0.0	0.0	0.0
Fe <sub>2</sub> O <sub>3</sub>	0.0	0.0	0.0	0.0	0.0	0.4	0.0	0.0	0.0	0.0	0.0	0.0	0.8	0.0	0.0	0.0
(Fe <sub>0.324</sub> Si <sub>0.676</sub> ) (Fe <sub>0.963</sub> Si <sub>0.037</sub> ) <sub>2</sub> O <sub>4</sub>	0.6	0.0	0.0	0.0	0.0	0.0	7.1	5.7	2.5	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca <sub>0.65</sub> Na <sub>0.32</sub> (Al <sub>1.62</sub> Si <sub>2.38</sub> O <sub>8</sub> )	0.0	0.0	0.0	0.0	14.3	69.8	64.1	73.6	64.3	0.0	0.0	4.8	0.0	0.0	0.0	0.0
Mg <sub>3</sub> Si <sub>2</sub> O <sub>5</sub> (OH) <sub>4</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.7	0.0	0.0	0.0	0.0	0.0
MgAlCrO <sub>4</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
KAl(Mg <sub>0.2</sub> Al <sub>0.8</sub> ) (Al <sub>0.42</sub> Si <sub>3.58</sub> )O <sub>10</sub> (OH) <sub>2</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.9	0.0	0.0	0.0	0.0	0.0
Na <sub>6</sub> K <sub>1.2</sub> Al <sub>7.2</sub> Si <sub>8.8</sub> O <sub>32</sub>	0.0	0.0	0.0	0.0	0.0	7.0	8.6	0.0	7.3	0.0	0.0	0.0	0.0	0.0	0.0	0.0
CaAl <sub>2</sub> Si <sub>2</sub> O <sub>8</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.5	0.0	0.0	0.0
SiO <sub>2</sub>	4.6	1.3	6.3	4.1	0.0	0.0	0.0	0.0	0.0	3.4	2.6	6.0	5.7	0.6	0.0	0.0
TiO <sub>2</sub>	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.6	0.1	0.0	0.0
Na <sub>1.71</sub> Al <sub>0.43</sub> (PO <sub>4</sub> )	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	1.8	0.0	0.0	0.0	0.0
Amorphous	87.2	97.2	93.7	90.1	79.0	4.1	10.1	19.0	22.7	96.5	95.0	78.5	72.5	99.1	100.0	100.0

SiO<sub>2</sub> and Al<sub>2</sub>O<sub>3</sub>. These samples are also depleted in CaO. Scrubber samples from tests T4 and T5 have ash compositions which are closer to the feed ash composition. However, all scrubber samples contain Cr<sub>2</sub>O<sub>3</sub> which was not present in the feed ashes. The Cr<sub>2</sub>O<sub>3</sub> must have originated from the gasifier refractory. The inorganic matter in the scrubber samples is 87-97% amorphous.

Table 8.7 provides surface area, skeletal density, envelope density and porosity of the scrubber samples. The scrubber sample from test T2 only has skeletal density provided since attempts to obtain nitrogen adsorption isotherms failed. Surface area of test T1 and T4 scrubber samples are approximately three times greater than the surface area of the test T5 scrubber sample. The T5 scrubber sample also has much lower porosity than the other scrubber samples, which is expected with lower surface area. It may be that this sample has less char, denser char and/or more plugging of pores by inorganic matter compared to the other scrubber samples. The T4 scrubber sample has the highest skeletal and envelope densities. This is expected since it has higher ash content than the other scrubber samples and inorganic matter is generally denser than organic matter. SEM images of the scrubber and probe samples are shown in Figure 8.4. Many features observed in these and additional SEM images were analysed by EDX. Annotations are added in Figure 8.4 to highlight some of these features. In the SEM images, char particles are darker than inorganic matter. EDX analyses of char surfaces without inorganic matter indicate a carbon content exceeding 80 wt.%. The structure of the char affects coalescence and expulsion of included particles of inorganic matter [11,12]. Yu *et al.* [13] present an equation which correlates the number percentage of char with more than 80% porosity ( $N_c$ ) to pressure ( $P$  in MPa) and the reactive maceral content of coal ( $M_r$  in %):

$$N_c = 6 \times P + 0.53 \times M_r + 37$$

**Table 8.7: Surface area, skeletal density, envelope density and porosity of solid samples**

<b>Test</b>	<b>Source</b>	<b>Surface area (m<sup>2</sup>/g)</b>	<b>Skeletal density (g/cm<sup>3</sup>)</b>	<b>Envelope density (g/cm<sup>3</sup>)</b>	<b>Porosity</b>
T1	scrubber	31.27	1.78	1.35	0.239
T2	scrubber	N/A	1.43	N/A	N/A
T4	scrubber	34.63	2.45	2.02	0.172
T5	scrubber	10.53	1.64	1.57	0.041
T2	probe	15.54	N/A	N/A	N/A
T3	probe	11.37	N/A	N/A	N/A
T4	probe	12.23	N/A	N/A	N/A
T1	tap	N/A	2.64	2.1	0.21
T2	tap	N/A	2.82	2.0	0.28
T3	tap	N/A	2.89	2.1	0.26
T4	tap	N/A	2.67	2.1	0.22
T5	tap	N/A	2.85	2.2	0.22

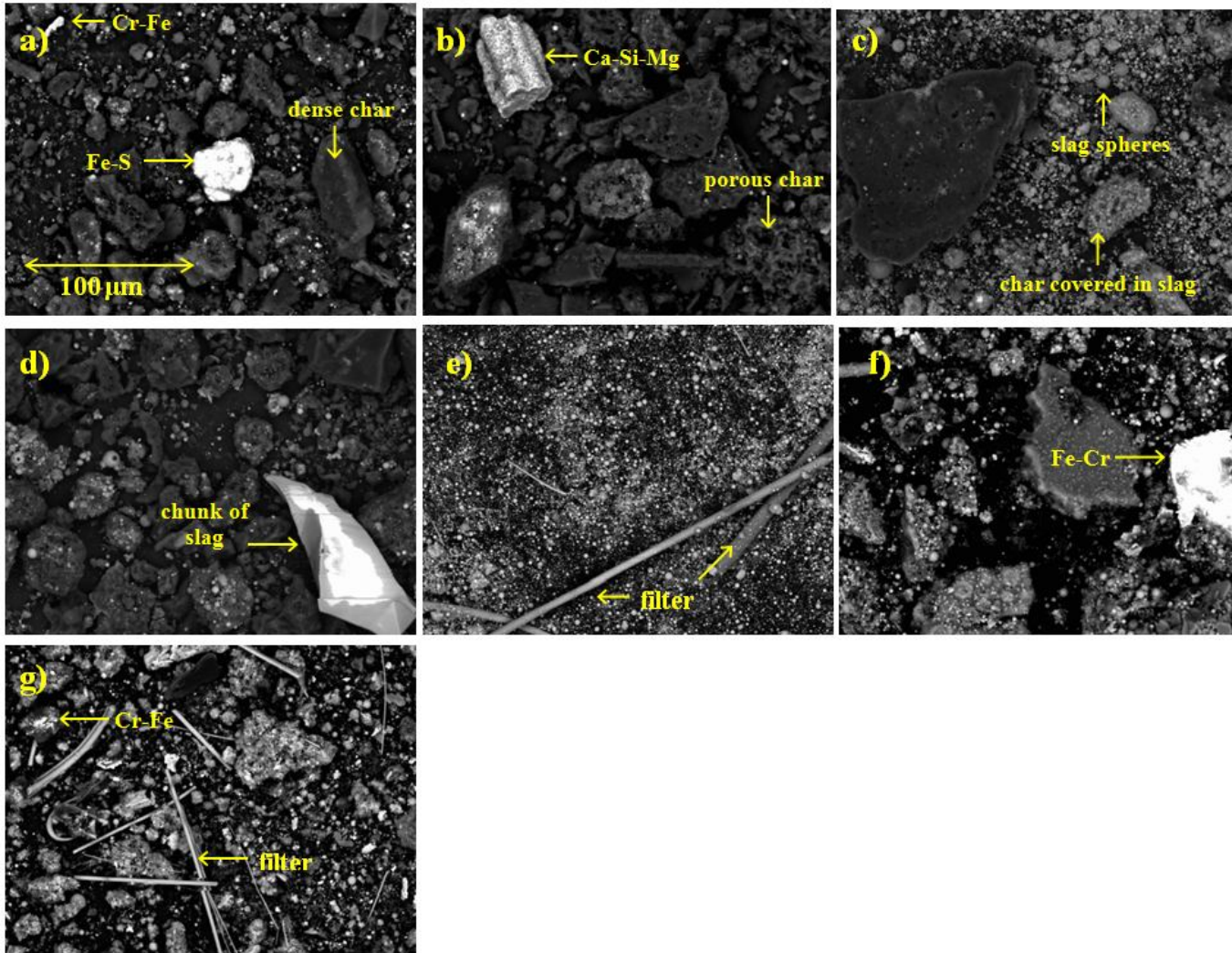


Figure 8.4: SEM images comparing scrubber and probe impinger samples: (a) T1 scrubber, (b) T2 scrubber, (c) T4 scrubber, (d) T5 scrubber, (e) T1 impinger, (f) T3 impinger and (g) T4 impinger. The scale in the (a) image applies to all images.

With the maceral content of the fuels obtained in the fuel characterization study [4], 88-92% of char particles are expected to have a porosity exceeding 80%. From SEM analysis of the scrubber and impinger samples, the proportion of highly porous char seems to be much lower than the predicted proportions. This may be due to higher conversion of porous char and differences in particle trajectories or sticking.

In all scrubber samples, most inorganic matter is spherical. Considering the low crystallinity of the ash in these samples (3-13%) and the high crystallinity of the feed ashes (71-95%), it appears that the majority of the inorganic matter melted in the gasifier and formed spheres due to surface energy minimization. Most of the inorganic matter on the char particles is spherical, indicating that the liquid slag on the char is non-wetting. The melting of inorganic matter and low wetting of char confirm assumptions made in ash particle formation models [11,12]. The compositions of the slag spheres are quite variable, but for the most part resemble the fuel ash composition. Many of the scrubber and probe samples contain separate non-spherical phases rich in chrome and iron. This indicates the chrome from the refractory material reacted with iron in the slag and spalling and/or erosion occurred. The strands observed in the probe samples are from the filters used in the probe impingers. Except for samples from test T4, the probe samples seem to contain less char than their scrubber counterparts. This result was validated by the measured carbon content of the samples as presented in Table 8.2. The discrepancies of char-to-ash ratios in the probe and scrubber samples may be caused by differences in particle trajectories and/or sticking to surfaces.

The carbon content of the scrubber samples is greater than the average carbon content of all solid samples combined. This suggests that compared to the proportion of fly ash particles collected from the scrubber, a greater proportion of char particles are entrained with the syngas to the scrubber rather than captured by water in the slag pot or quench discharge. For particles to make it to the scrubber, they must flow down with the gas phase through the slag tap and into the quench vessel, and then be brought up with the gas to the syngas outlet instead of submerging into the water in the quench vessel. According to Stokes's law, it is easiest for smaller and lower density particles to achieve

this feat. Char particles have lower density than fly ash particles and are therefore likelier to be entrained. There are several possible causes to explain why a greater amount of ash made it to the scrubber during test T4. The fly ash could have been smaller or had a lower density. Also, the gas velocity may have been greater. Alternatively, the fly ash particles may have had a lower wall impaction/sticking rate with the surfaces in the gasifier during test T4.

#### **8.4.3 Slag from the slag tap, slag pot and quench water discharge**

Major and minor oxides compositions of samples collected at the slag tap, in the slag pot and in the quench water discharge are provided in Table 8.5. All slags from a given test have similar major and minor oxides compositions. Crystalline phase content of samples collected at the slag tap and in the slag pot is provided in Table 8.6. Skeletal density, envelope density and porosity of slag tap samples are provided in Table 8.7.

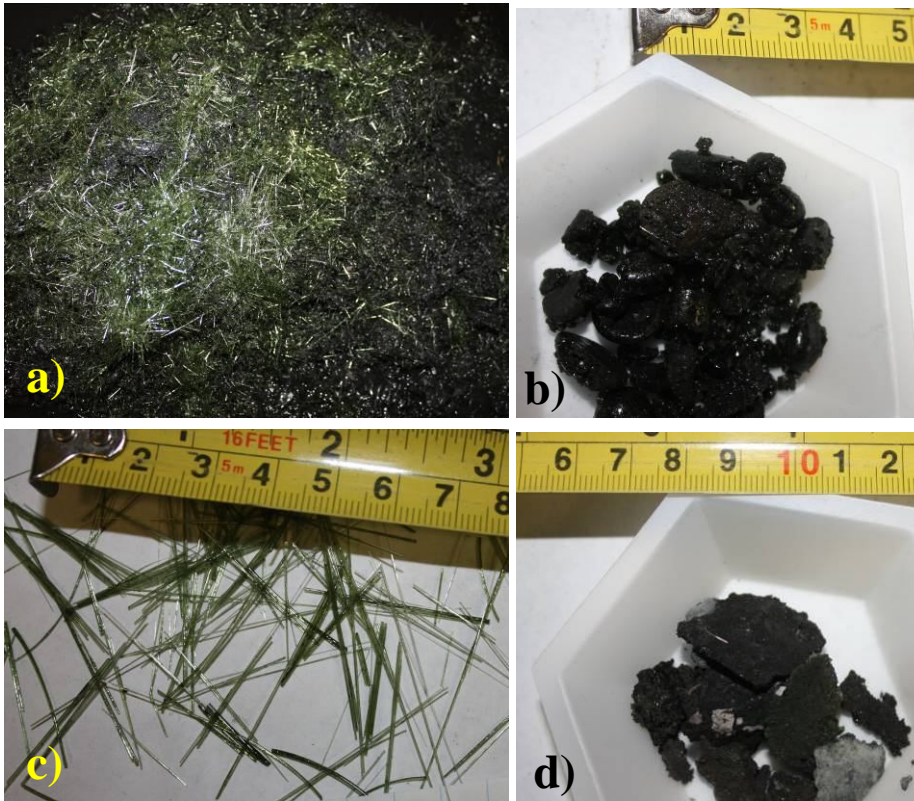
Together,  $\text{SO}_3$  and  $\text{Na}_2\text{O}$  represent 4-20 wt.% of the feed ashes. However, they are expected to partially or fully devolatilize in the gasifier. Compared to the feed ashes, the content of  $\text{SO}_3$  in the slag tap, slag pot and quench water discharge samples decreased by 65-100%. The content of  $\text{Na}_2\text{O}$  decreased by 15-52% in all these samples except test T4 samples which have increases of 123-203%, and the quench water discharge sample from test T3 which has an increase of 7%. The increase in sodium content for test T4 slags is likely due to contamination from slag accumulated in the gasifier during previous tests. Nonetheless, the amount of  $\text{Na}_2\text{O}$  in these slags remains small ( $\leq 1$  wt.%). Excluding the  $\text{SO}_3$  and  $\text{Na}_2\text{O}$  loss, most slag samples are enriched in  $\text{SiO}_2$ , enriched in  $\text{Al}_2\text{O}_3$  and depleted in  $\text{CaO}$ . These changes may be caused by interactions with the gasifier refractory and slag accumulated in the gasifier during previous tests.

$\text{Ca}_{0.65}\text{Na}_{0.32}(\text{Al}_{1.62}\text{Si}_{2.38}\text{O}_8)$  dominates the crystalline portion of the slag tap slags. This compound has a composition which is similar to the slag major and minor oxides compositions and has a melting temperature of 1400 °C [14]. The slag tap slags also contained up to 3.2 wt.%  $\text{FeCr}_2\text{O}_4$ , which has a melting temperature of 1670 °C [15]. The

$\text{FeCr}_2\text{O}_4$  in the slag at the slag tap is likely a result of iron in slag reacting with the refractory and forming  $\text{FeCr}_2\text{O}_4$  which then detaches from the refractory, or chromium from the refractory dissolving in slag and then precipitating as  $\text{FeCr}_2\text{O}_4$ .

The slag pot samples have much higher amorphous content than the slag tap samples (Table 8.6). The higher amorphous content of the slag in the slag pot suggests that most of the slag flowed down from the slag tap and was rapidly quenched. Solids with different shapes were discovered in the slag pot, as can be seen in Figure 8.5. Slag chunks were found in all the slag pot samples. Their major and minor oxides composition is roughly the same as the slag pot slag composition (Table 8.5). However, they contain much less carbon (Table 8.2). These chunks are slag which flowed from the slag tap and into the quench vessel. Filaments were found in slag pot samples from tests T1, T3 and T5. These are mostly translucent and green. They are completely amorphous (Table 8.6) and their major and minor oxides composition is roughly the same as the slag pot slag composition (Table 8.5), although they contain almost no carbon (Table 8.2). Slag sheets were found in slag pot samples from tests T1 and T5. They contain very little carbon (Table 8.2). Slag sheets from test T1 resembled the slag pot slag composition in terms of major and minor oxides (Table 8.5). However, sheets from test T5 are highly enriched in  $\text{Al}_2\text{O}_3$  and  $\text{Cr}_2\text{O}_3$ . The sheets from test T1 may be cracked pieces of slag. Sheets from test T5 seem to be pieces of slag with refractory.

The carbon content of slag samples from the quench discharge water (9-51%) is higher than the carbon content of slag from the slag pot (3-26%). Hence, the proportion of char that goes in the quench water discharge (as opposed to remaining in the slag pot) is greater than the proportion of ash which goes in the quench water discharge. This is expected since the higher density slag can rapidly sink down in the water in the slag pot, while lighter char tends to stay on top or does not sink as quickly. The water which is removed from the top of the slag pot can therefore entrain a higher fraction of the char in the higher slag pot water as it is discharged. The quench water does contain a significant portion of the unconverted char particles (Table 8.4). Nonetheless, for four of the five tests, more carbon was present in the slag pot.



**Figure 8.5: Slag collected from the slag pot after test T5 (a) and classified according to shape: (b) slag chunks, (c) slag filaments and (d) slag sheets.**

Beyond the atomic composition of slag, there is much interest in knowing the oxidation state of slag iron in the gasifier since it can affect slag flow and interaction with the refractory. The oxidation state of the iron in the slag will depend on many factors such as temperature, gas phase composition and slag composition [16]. Assumptions on the oxidation state can be made based on temperature and oxygen partial pressure in the gasifier, but whether gas-slag equilibrium is achieved remains unclear.  $^{57}\text{Fe}$  Mössbauer spectroscopy measurements indicate that virtually all the iron atoms in the slag collected from the slag pot after tests T1-T5 are in the divalent oxidation state. This suggests that iron in slag flowing through the slag tap is in the divalent oxidation state before reaching the quench vessel and experiencing rapid quenching.

#### 8.4.4 Slag layer on refractory

The following equation to calculate the thickness of slag ( $\delta$ ) assumes a steady-state falling film of slag on a vertical surface with laminar flow, negligible rippling at the slag-gas interface, no slip at the wall, slag entrainment is entirely caused by gravity (i.e., there is no shear stress at the slag gas interface), all ash entering the gasifier sticks to the wall, the film thickness is much smaller than the reactor diameter, and the slag properties are constant:

$$\delta = \left( \frac{3\dot{m}\mu}{\rho^2 g C} \right)^{1/3}$$

where  $\dot{m}$  is the injection rate of ash which is calculated from slurry used, test duration, slurry moisture content and fuel ash content.  $\rho$  is the density of the slag which is taken as the envelope density of slag at the slag tap.  $g$  is acceleration due to gravity.  $C$  is the circumference of the refractory liner.  $\mu$  is the slag viscosity. For slag thickness predictions with the upper refractory liners, the TC1 temperature is assumed. For slag thickness predictions with the probe sheaths, the TC4 temperature is assumed. For each sheath and liner, the slag thickness calculations were first completed by predicting the slag viscosity from measured values in the fuel characterization study [4]. This assumes the feed ash composition applies. The thickness calculations were then repeated with the assumption that the slag pot ash composition applies. In this case, the slag viscosity model used for prediction was selected using the method presented by Duchesne et al. [17]. Ash injection rates, predicted slag viscosities and predicted slag layer thicknesses are given in Table 8.8. The slag viscosities predicted with the slag pot slag composition tend to be higher than those predicted with the feed ash composition. In some cases, the former is more than five times greater than the latter. Still, the two thicknesses never differ by more than 1.5 mm.

The slag flow in the gasifier can be considered laminar if the Reynolds number ( $N_{Re}$ ) is less than 1200 and negligible rippling will occur when it is lower than 20 [18,19]. The Reynolds number can be obtained from:

$$N_{Re} = \frac{4\rho^2\delta^3g}{3\mu^2}$$

The calculated Reynolds numbers for slag flowing on the upper refractory liner and on the probe sheath are given in Table 8.8. Values for each predicted slag viscosity are given. The Reynolds numbers are very small and the assumption of laminar flow with negligible rippling is applicable. In all cases, with all other parameters constant, the slag thickness would have to be at least ten times greater for non-negligible rippling to occur. Slag thicknesses on the liners and probe sheaths were measured using Vernier calipers. For the liners, averages of ten measurements with standard deviations are provided in Table 8.8. For the probe sheath, the maximum thickness is provided in Table 8.8.

Figure 8.6 shows photographs of upper refractory liners collected after each test. The measured slag surface roughness values are shown in Figure 8.7. Liners from tests T1 and T2 have smoother slag surfaces than the liners from tests T3-T5. Slag features on liners from tests T1 and T2 are less than 0.5 mm thick. Slag features on liners from tests T3-T5 are up to 12 mm thick, but for the most part are much thinner. The predicted thicknesses for liners from tests T1 and T2 are greater than what is observed experimentally. The predicted thicknesses for liners from tests T3-T5 are smaller than what is observed. The liner from test T1 presents some slag rivulets which are 3-50 mm in width and less than 0.5 mm thick. Some rivulets travelled the entire length of the liner. T2 also presents some slag rivulets, however they are less numerous than for the liner from test T1. They are 3-20 mm wide, 10-200 mm long and less than 0.5 mm thick. Most of the slag on the T2 liner is present as spots with diameters ranging from 1 to 80 mm. The larger spots are mostly near seams, cracks and holes in the liner. The slag layers on liners from tests T3-T5 are rough and bubbly. Bumps can protrude up to 35 mm. Some cavities are as deep as 20 mm. Gas bubbles may have been formed by decomposition or devolatilization of slag

components. Reaction with the refractory material may have prompted bubble formation as well. Another hypothesis is that gas trapped in the refractory pores escapes as bubbles while the slag penetrates the refractory material. Clearly, assumptions made in the above slag thickness model do not represent reality and the formation of rivulets and bubbles warrants future studies.

**Table 8.8: Ash injection rates and slag layer properties**

<b>Test</b>	<b>T1</b>	<b>T2</b>	<b>T3</b>	<b>T4</b>	<b>T5</b>
<b>Ash injection rate (kg/h)</b>	5.0	3.9	4.1	4.4	10.7
<b>At upper liner</b>					
<b>Viscosity<sup>a</sup> (Pa·s)</b>	3.9	2.1	1.9	1.7	1.9
<b>Viscosity<sup>b</sup> (Pa·s)</b>	12.6	4.6	3.1	1.4	2.8
<b>Model slag thickness<sup>a</sup> (mm)</b>	0.8	0.6	0.6	0.6	0.8
<b>Model slag thickness<sup>b</sup> (mm)</b>	1.2	0.8	0.7	0.6	0.9
<b>Measured thickness<sup>c</sup> (mm)</b>	<0.5	<0.5	2.7 (1.2)	3.0 (0.6)	4.2 (3.5)
<b>Reynolds number<sup>a</sup></b>	0.0018	0.0026	0.0038	0.0046	0.0100
<b>Reynolds number<sup>b</sup></b>	0.0006	0.0012	0.0024	0.0056	0.0068
<b>At probe sheath</b>					
<b>Viscosity<sup>a</sup> (Pa·s)</b>	16.9	27.4	27.8	4.6	10.9
<b>Viscosity<sup>b</sup> (Pa·s)</b>	124	240	173	6.9	19.8
<b>Model slag thickness<sup>a</sup> (mm)</b>	1.3	1.4	1.5	0.9	1.5
<b>Model slag thickness<sup>b</sup> (mm)</b>	2.5	2.9	2.8	1.0	1.8
<b>Measured thickness<sup>d</sup> (mm)</b>	19.0	4.0	<0.5	4	14.0
<b>Reynolds number<sup>a</sup></b>	0.00042	0.00020	0.00026	0.00169	0.00174
<b>Reynolds number<sup>b</sup></b>	0.00006	0.00002	0.00004	0.00113	0.00096

<sup>a</sup> assuming feed ash composition

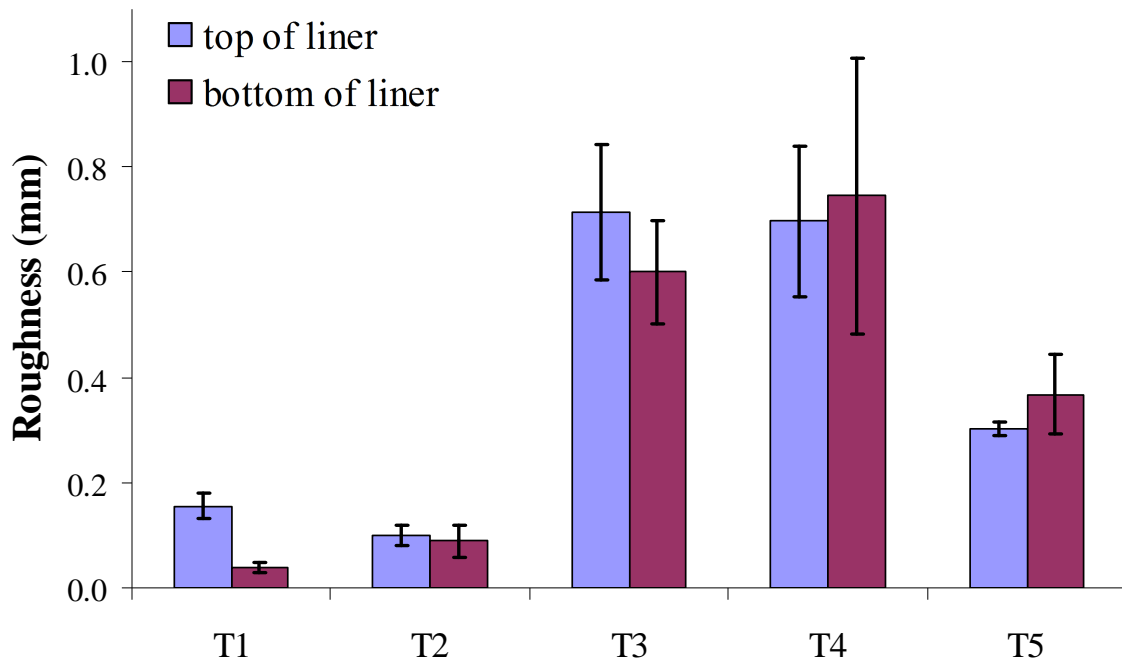
<sup>b</sup> assuming slag pot slag composition

<sup>c</sup> standard deviation is in brackets

<sup>d</sup> maximum thickness on sheath



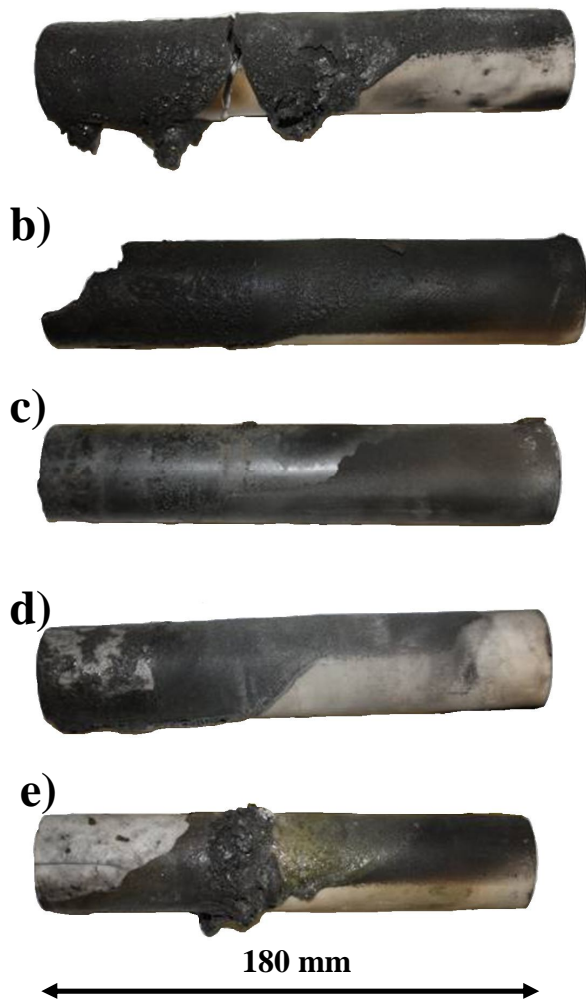
Figure 8.6: Photos of liners (a-e) from tests T1-T5, respectively.



**Figure 8.7: Roughness of slag surface on liners. The averages of five measurements are given. Variability bars represent standard deviations.**

Figure 8.8 shows photographs of probe sheaths collected after each test. One reason for collecting probe sheaths from each gasification test is to provide validation for models describing char and ash particles contacting surfaces in the gasifier. For larger particles ( $>1 \mu\text{m}$ ), inertial impaction is the dominant mechanism [20,21]. Momentum normal to the surface, gained by particles from gas entrainment away from the surface or from turbulence, must overcome the gas stream lines near the surface. The particle impaction efficiency can be correlated to the Stokes number [22]. The correlation for a sphere approaching a cylinder in cross flow can provide reasonable accuracy for this purpose. Correlations relating impaction efficiency to the Stokes number are available for parallel flow, but their validity is less certain than for inertial impaction in cross-flow [22]. Hence a cylindrical probe sheath placed in cross-flow which can be rapidly quenched by retraction from the gasifier provides better validation data for impaction models than slag collected from a wall with parallel flow and slower quenching. In the following, the

convention used refers to the left of the probe sheath as the end that is in the gasifier. The top corresponds to the top of the sheath as it was placed in the gasifier. All probe sheaths are made of alumina, except for the sheath used in test T3 which is made of silicon carbide. The probe sheath from test T1 cracked at 70 mm from the left end. Its entire top is covered in slag which is typically less than 1 mm thick. Near where it cracked, slag is up to 5 mm thick. The section of the sheath on the right that did not extend into the gasifier can be distinguished as it does not have slag at the bottom. The left section which was extended past the gasifier inner refractory has dripping slag, up to 19 mm thick. As for the probe sheath from test T2, a piece broke off the top left. The entire top and left side are covered in slag, generally less than 0.5 mm thick. The top left has areas with thicker slag, but still less than 1 mm thick. The left bottom has slag which is up to 4 mm thick. The bottom of the right section which did not extend into the gasifier is bare. For the silicon carbide sheath collected from test T3, slag on its left side was knocked off as it was removed from the gasifier. Some slag remained on the top right side. The built-up slag thickness is less than 0.5 mm. Probe sheaths from tests T4 and T5 both had some slag flake off the top left sections. The sheath from test T5 has more slag on it, particularly on the top section aligned with the inner gasifier refractory. The measured maximum thicknesses on the probe sheaths are compared to predicted values in Table 8.8. For all tests except T3, the predicted thicknesses are smaller than the maximum measured thickness. Greater predicted thicknesses do not correspond to greater measured maximum thicknesses. The slag thickness model is a poor representation of the slag layer on the sheaths. A better model for slag flow on the sheaths would likely consider interfacial tension. For example, less slag may have accumulated on the silicon carbide sheath than the alumina sheath due to poorer sticking of char/ash/slag particles during impaction on silicon carbide. Sticking is a function of the gas-slag, gas-refractory and slag-refractory interfacial tensions [23]. Also, the bare silicon carbide sheath is smoother than the bare alumina sheaths. Generally, wetting on rough surfaces is greater than on smooth surface when the contact angle between the liquid and solid is less than  $90^\circ$  [10].



**Figure 8.8: Photos of probe sheaths (a-e) from tests T1-T5, respectively.**

The section on top of sheaths with thicker slag was aligned with the wall during gasifier testing. This suggests slag flowed down from the refractory and onto the sheaths. Another possible cause of the thicker slag would be higher char and fly ash particle density in the gas phase near the wall, leading to increased impaction on the sheath in this area. This can occur if particle sticking on the refractory is inhibited. Such is the case if there is extensive coverage of char on the surface of slag flowing down the refractory [24,25].

#### **8.4.5 Slag-refractory interface**

Pieces of slag-covered refractory near the mid-section of the upper refractory liner were cut and cross-sections were used for SEM-EDX analysis. Temperature during gasifier operation at the slag surface of these sections was assumed to be near the TC1 temperature given in Table 8.1. Figure 8.9 shows silicon, aluminum and calcium elemental maps at the slag-refractory interface for tests T1 and T2. The interface between the slag and alumina is clearly seen since little slag penetrated the alumina which has high aluminum and low silicon concentrations. Spalling/eroding of alumina is seen in Figure 8.9. SEM-EDX results for the slag near the alumina refractory indicate a 73% and 88% increase of atomic aluminum content for tests T1 and T2, respectively. The distribution of elements in the slag suggests most of the slag has formed crystal plagioclase with a small portion of amorphous slag surrounding the crystals. It is not clear whether these crystals were present during the test or only formed upon shutdown and cooling of the gasifier system. SEM images indicate that the slag layer thickness varies but is typically 0.2 mm thick.

Percentages of slag penetration into the refractory were calculated as follows. The atomic composition of a given refractory section was calculated by averaging the results of a minimum of 20 SEM-EDX spectra. An estimated atomic composition of this section was then expressed as a weighted average atomic composition combining the test's feed ash composition and the slag-free refractory composition. The sum of squared differences between the components in this weighted average and the SEM-EDX spectra average was minimized by allowing the weight of the feed ash and slag-free refractory to vary from 0 to 1, with their sum equal to 1. The weight of the feed ash is the slag penetration fraction. At a depth of 0 to 150  $\mu\text{m}$  into the refractory, in reference to the slag-refractory interface, samples from tests T1 and T2 have 3.5% and 5.4% slag penetration, respectively.

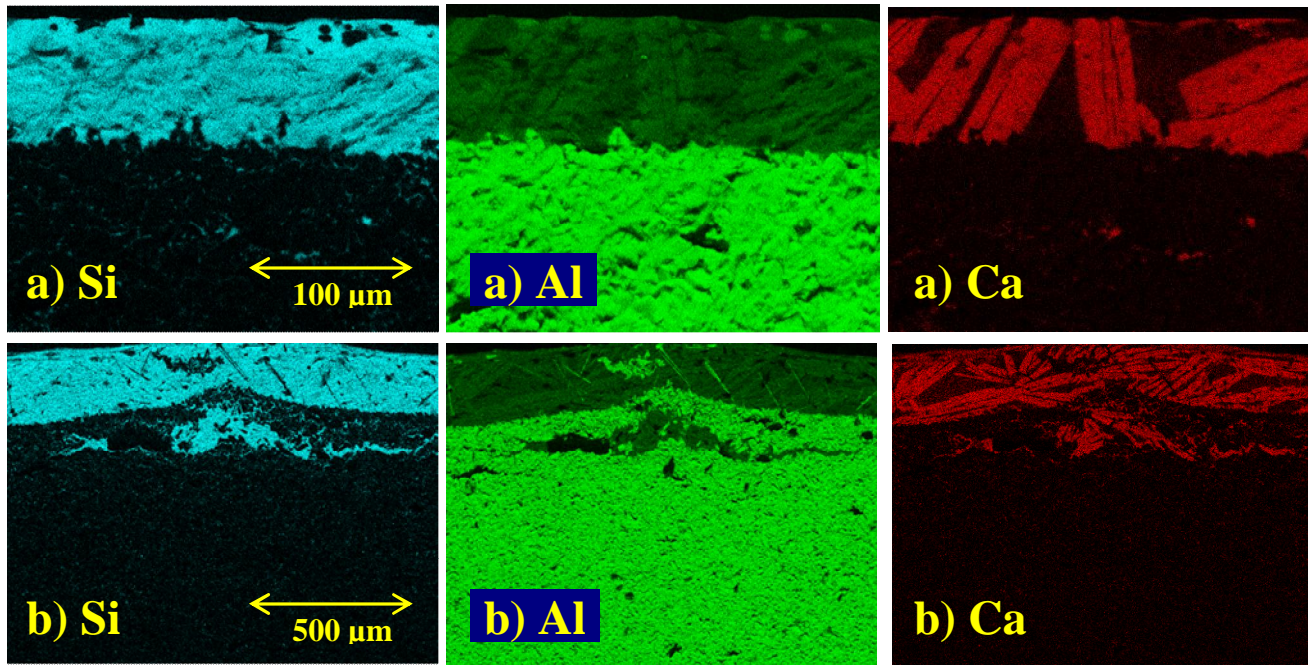


Figure 8.9: Silicon (Si), aluminum (Al) and calcium (Ca) elemental maps produced by SEM-EDX of slag-refractory cross-sections for tests T1 (a) and T2 (b). The scale in the left image applies to the centre and right images.

Elemental maps for tests T3-T5 are not shown since it is difficult to locate a slag-refractory interface. Slag penetration was calculated as a function of depth into the refractory, in reference to the slag-covered surface. Results are given in Figure 8.10. Slag penetrated approximately 20, 12 and 8 mm for tests T3, T4 and T5 respectively. Even in the first millimeter of depth, where the sample looks like dark slag, the purity of the slag is less than 80%. The slag percentage does not continuously decrease with depth. This can be due to the presence of pores, cracks in the refractory, or due to the heterogeneous composition of the refractory. Slag penetration is a major cause of refractory wear. Penetration depth is, among other factors, a function of temperature, slag viscosity, slag-refractory interfacial tension, slag-refractory reactions, refractory porosity and exposure duration [26]. The viscosity of slag on the liners should be similar for tests T3-T5, as indicated in Table 8.8. However, temperature, exposure time and slag compositions vary between tests and it is difficult to isolate which parameters had the greatest effect on penetration.

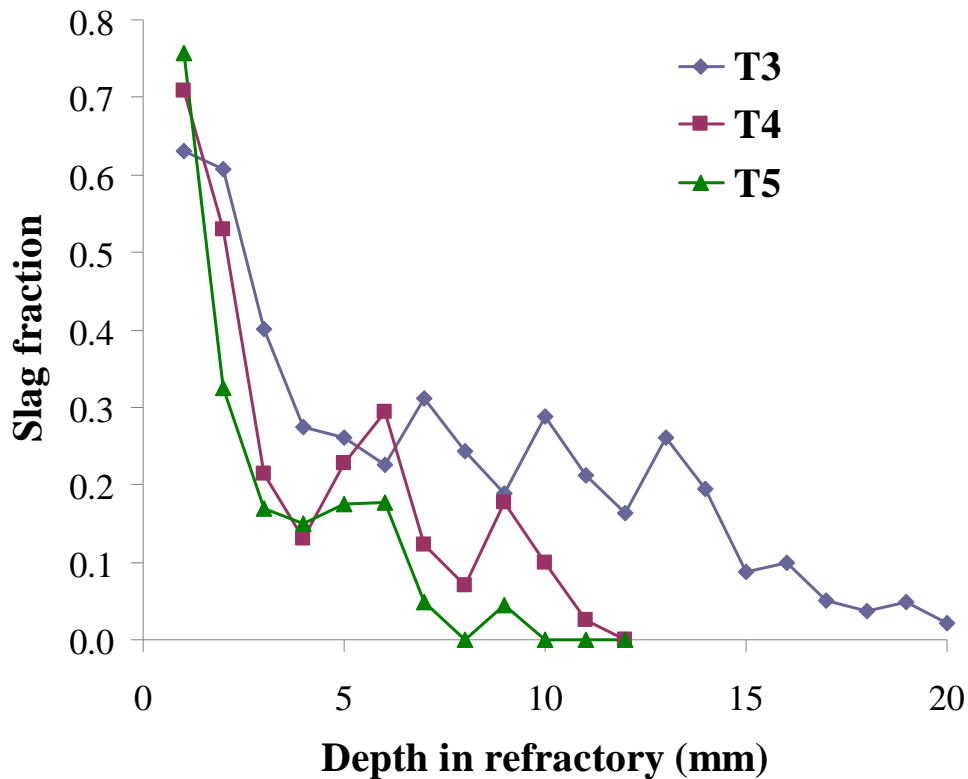


Figure 8.10: Slag penetration in alumina-chromia refractory.

## 8.5 Conclusion

Five tests have been completed with CanmetENERGY's pilot-scale gasifier. The first three tests were with fuel F1, the fourth with F3:L2 and the fifth with F4:L2. The gasifier configuration allowed replacement of the upper refractory liner after each test. Alumina liners were used for the first two tests. Alumina-chromia liners were used for subsequent tests. Carbon conversions and cold gas efficiencies for the tests were within 97.0-99.6% and 33.7-61.9%, respectively. Samples collected from the gas probe impingers and scrubber have relatively high carbon concentrations, but most of the unconverted carbon was collected in slag from the slag pot and quench water discharge. Char and fly ash samples indicate that most of the inorganic matter melted and formed spheres, regardless of whether it was associated to char or not. Amorphous slag ran down the slag tap and dripped into the slag pot. Some slag accumulated near the slag tap and crystallized. Slag collected from the gasifier was affected by devolatilization and interaction with the gasifier refractory. This can significantly impact slag properties such as viscosity. Signs of spalling and/or erosion were detected by SEM-EDX analysis of the slag-refractory interface from the first two tests. The chromium-rich particles found in scrubber, probe impinger and slag pot samples are other signs of spalling/erosion. Slag accumulated on the alumina liners is thin (~0.2 mm) and formed rivulets and spots. The slag is enriched in aluminum and penetration into the alumina refractory was 3.5-5.4% at a depth of 0-150  $\mu\text{m}$ . Slag layers on the alumina-chromia liners are rough and bubbly. For these liners, the slag-refractory interface is difficult to distinguish, but slag thickness was estimated at 2.7-4.2 mm. Slag penetrated 8-20 mm into the alumina-chromia liners. Measured slag thicknesses were compared to predicted slag thicknesses. The slag thickness model used should be improved by considering interfacial tensions, deviation from the vertical wall assumption and non-constant slag properties.

## 8.6 Acknowledgements

Jeffery Slater and Ryan Burchat of CanmetENERGY were invaluable for the operation and maintenance of the pilot-scale gasifier. Part of this research has been funded by the Natural Sciences and Engineering Research Council of Canada (NSERC) and the Canadian Federal Program for Energy Research and Development (PERD).

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## Chapter 9. Conclusions and recommendations

### 9.1 Slag viscosity measurements

The viscosities of slags have been widely investigated over the past five decades. Slag viscosity has major implications in glass production, geophysical phenomena, metallurgical processes, combustion and gasification. Measurements have been conducted with samples from different origins, including artificial laboratory prepared glasses and ashes, coal fly ashes, magmatic systems, gasifier slags, and metallurgical slags [1-6]. In the case of entrained-flow slagging gasification, slag viscosity plays a role in ash particle sticking, slag flow and slag-refractory interactions. More measurements are required for slags produced from gasification fuels that are of interest in Canada and abroad, such as petroleum coke (petcoke).

Petcoke slags contain important amounts of vanadium which is typically not found in other types of slag. Vanadium's multiple oxidation states and complex phase formations make it challenging to predict its interaction with the other components of slag [7,8]. In Chapter 5, it is determined from viscosity measurements and microstructural analyses that adding petroleum coke ash to coal ashes changes the relative amounts of network formers and modifiers, as well as the liquidus temperature and types/quantity of solids in the resulting slags. Vanadium has the particular effect of promoting the formation of spinel in slags below 1300 °C. However, it was also noted that vanadium-rich slags ( $\geq 5$  wt%  $V_2O_5$ ) require development of a better viscosity measurement technique due to high reactivity with spindle/crucible materials. Also noteworthy from Chapter 5 is the repeatability of results with the CanmetENERGY viscosity measuring apparatus, and the reproducibility of results between the CanmetENERGY and CSIRO viscosity apparatuses.

In Chapter 6, the interactions between vanadium-rich slags and support materials used for viscosity measurements were investigated in more detail. The bulk and phase compositions of two petcoke slags on  $\text{Al}_2\text{O}_3$ , Mo, Pt and Ni supports produced under different laboratory conditions were analysed, and kinetics of slag composition changes at 1400 °C were determined. Interactions of slag and support materials lead to changes in bulk slag composition and appearance of new solid phases. An interface layer between the slag and support was formed and identified as V-rich oxide, Mo-V-oxide and spinel for slags processed with Ni, Mo and  $\text{Al}_2\text{O}_3$  respectively. The interface layer can act as a barrier for migration of the slag components to the supports and elements of the support to the slags. Viscosity of slags with Mo, Ni and  $\text{Al}_2\text{O}_3$  supports were determined in the temperature range 1200-1500 °C. Interactions with the support materials affected the bulk composition of the processed slag and as a result impacted viscosity. It was determined that Ni and Pt supports can be used for viscosity tests in conditions where formation of metallic Fe and Ni can be minimised or avoided. Mo supports can be used for viscosity measurements where the total concentration of  $\text{V}_2\text{O}_3$  in the petcoke (or coal-petcoke blend) ash is less than 5 wt.%.  $\text{Al}_2\text{O}_3$  supports can be used for slags with higher than 5 wt.%  $\text{V}_2\text{O}_3$ . However, when using  $\text{Al}_2\text{O}_3$  supports, the starting mixture has to be modified in order to compensate for changes in Al and V concentrations in the slag during viscosity tests.

As was mentioned in previous chapters, the main parameters that affect slag viscosity are composition and temperature. The exact composition may not be known due to changes in oxidation/coordination states of the components. For instance, the most studied effect of gas atmosphere on slag viscosity relates to the oxidation state of iron [9-13]. Iron can be present in slags as ferric iron ( $\text{Fe}^{3+}$ ), ferrous iron ( $\text{Fe}^{2+}$ ) or metallic iron ( $\text{Fe}^0$ ). Generally, ferric iron increases the viscosity and ferrous iron decreases the viscosity. The effect of metallic iron is less clear since in most slag viscosity measuring systems the metallic iron forms a separate phase which accumulates at the bottom of the sample. Iron is not the only slag component with multiple oxidation/coordination states. Aluminum, vanadium, chromium, titanium and manganese can also have multiple oxidation/coordination states. Aluminum, a common major component of gasifier slags,

is known to have various coordination states [14,15]. Aluminum tends to increase viscosity when its charge can be balanced by alkali and alkaline earth oxide metals [16]. Otherwise, it tends to decrease viscosity. Vanadium is a major component of petcoke slags. Its effect on slag viscosity has already been discussed in Chapters 5 and 6. With more experimental phase equilibria data, vanadium could be added to the slag solution database in FactSage, allowing much improved phase equilibria predictions for petcoke slags. The concentration of chromium is usually negligible in gasifier feedstocks. However, as was discussed in Chapter 8, chromium is often used in gasifier refractory materials and can be incorporated into slag flowing down gasifier walls. Viscosity of slags containing chromium has not been studied systematically and further research is needed [17]. Titanium and manganese generally have a low prevalence in gasifier slags.

In addition to changes in oxidation/coordination of slag components, dissolution and devolatilization may affect the composition of slag. There have been studies claiming  $H_2O$ ,  $H_2$ ,  $N_2$ ,  $NH_3$  and  $CO_2$  can dissolve in slag and significantly affect its viscosity [9,18-20]. However, predicted amounts of these dissolved components are very small [21]. More experimentation is required. The dissolution of carbon in slag is of particular interest for gasification. Furthermore, some slag components may volatilise into the gas phase (Chapter 8). Alkali metals typically have higher vapour pressures than other slag components and have been known to devolatilize [16]. Slag viscosity can increase due to their loss. The amount being volatilised is likely dictated in part by the surrounding atmosphere's composition. Undoubtedly, the kinetics of devolatilization must be considered when measuring the viscosity of slags with high concentrations of alkali metals or other volatile compounds. Composition analyses of such slags after viscosity measurements can be useful.

Although many aspects of viscosity measurements remain to be improved, one of the often unaddressed aspects is non-Newtonian behaviour. Slags tend to have non-Newtonian behaviour when they contain solids and/or two or more immiscible liquids [16]. Non-Newtonian behaviour can be detected by varying the shear rate during viscosity measurements and by monitoring viscosity over time. Many slag viscosity

studies justifiably avoid measurements with non-Newtonian flow since it is more complex to model and/or not necessarily of interest for their application. For instance, there is no need to measure non-Newtonian flow properties if the only application of the data will be to apply a maximum viscosity (or minimum temperature) heuristic for fully molten slag to properly flow through the slag tap in a gasifier. However, as the application of slag viscosity broadens (i.e., particle sticking probabilities and slag layer viscosity gradients in gasifiers) and the models improve (see Section 9.2), the need for non-Newtonian viscosity measurements increases. In one of the few studies providing non-Newtonian slag flow data, the viscosity of 45 coal ashes at 1200-1600 °C with varying shear rates was measured [22]. For the advancement of gasification modeling, more data considering the shear rate dependence of viscosity should be acquired.

## 9.2 Slag viscosity modeling

There is great interest in slag viscosity models since measurements are costly, lengthy to perform and require much expertise to obtain accurate data. Many slag viscosity models are described in literature, but their accuracy and range of applicability are limited. The work presented in Chapter 3 is an attempt to create a model with good accuracy and applicable to a broad range of temperatures and slag compositions. Artificial neural networks (ANNs) present significant advantages over conventional data regression models as there is no need to provide a model function, the learning capability of ANNs allows the inclusion of more complex and subtle interactions between variables, and ANNs are intrinsically more robust when it comes to handling noisy or inaccurate data [23]. An ANN was created to predict slag viscosity with eight components, temperatures ranging 809-2482 °C and viscosities ranging  $10^{-3.81}$ - $10^{11.16}$  Pa·s. It outperforms other models with an average absolute logarithmic error (AALE) of 0.703 when applied to data not used for its training. Despite being better than the other models, an AALE of 0.703 means that the predicted value is on average 5 times greater or smaller than the measured value.

Rather than seeking to create a one-size-fits-all model, a strategy to determine which empirical model applies best to certain slags and conditions can be applied. Prior to the work presented in Chapter 4, a method was sought to determine whether a particular slag composition and temperature are within the range of data used for a given model. This is not an easy task as the data is found in n-dimensional space, with n being the sum of the number of components considered plus one for temperature. However, if the data used to create a model are known, a linear program can be used to determine whether a prediction falls within the convex hull formed by that particular model's data [24]. However, the weakness of this extrapolation detection approach is that it does not consider that a model may have better performance with predictions that are outside of its convex hull but *close* to its data, compared to predictions which are within its convex hull but *far* from its data. An alternative to the convex hull strategy is to consider the data density in the model's n-dimensional space. Various tree-based density estimation strategies can achieve this [25]. Assuming the model predicts well for data used to create it, or similar data, a data density approach can be useful. Nevertheless, data density does not guarantee good performance since models do not necessarily perform well with all the data used to create them. A more direct strategy to assessing models is to apply them to the data in the region of n-dimensional data space which is of interest. For this, a large database increases the probability of having good data density in the region of interest. The consideration of numerous models increases the probability of finding one that performs well in the region of interest. The challenge lies in obtaining a large database, developing numerous models, and manipulating the database and models together.

In Chapter 4 a slag viscosity modeling toolbox, comprised of the SlagViscosityPredictor tool (SVP) and SlagViscosityDatabase tool (SVD), is presented. The toolbox was developed to facilitate the evaluation and selection of slag viscosity models for given slag compositions and conditions. The SVP includes a new model, ANNliq, for liquid slag viscosity prediction. It also can apply the Roscoe-Einstein (RE) model in conjunction with viscosity models and FactSage phase predictions. The recommended strategy is to first determine which slag compositions and conditions are of most interest. The SVD is

then used to identify viscosity data that match or are similar to these slag compositions and conditions. The SVD will also provide the number of viscosity measurements which match the selection criteria, and the performance of 24 slag viscosity models with these measurements. Provided there is sufficient data which match the selection criteria, the model with the best performance is noted. Then, the specific slag compositions and conditions are entered in the SVP to generate slag viscosity predictions with the best model. Model assessment with the toolbox should be done with great care to balance the stringency of criteria to identify data in the database and the quantity of data identified.

Emphasis should be placed on the unconventional performance function used in Chapters 3 and 4 to compare viscosity models. Conventional performance functions such as the average absolute error (AAE) and average absolute relative error (AARE) are inadequate for slag viscosity model evaluation. The AAE does not account for error tolerances which are proportional to the slag viscosity value. The AARE gives preference to predicted values which are proportionally smaller rather than proportionally larger. This becomes more apparent as the relative error becomes larger. In the case of slag viscosity predictions, relative error greater than 100% is common. Note that the average squared error and average squared relative error present the same problems as the AAE and AARE, respectively. The AALE always favours the predicted values which are proportionally closest to the measured values, regardless of whether the predicted values are smaller or greater than the measured values.

A promising slag viscosity model which was excluded from the study in Chapter 4 is the recently developed model that was built into FactSage [26]. FactSage 6.2 was used for the work in Chapter 4 and does not have macro commands to automate viscosity calculations. However, this feature is now available in FactSage 6.3. Future work will include adding the FactSage viscosity calculator to the Slag Viscosity Modeling toolbox.

With the case studies in Chapter 4, it was found that the RE-modified viscosity models did not have the best performances. This was unexpected since the RE-modified viscosity models account for solids in slag, while most other slag viscosity models are semi-

empirical and their theoretical basis is for fully molten slags. The ANNs could account for solids since they are fully empirical, but it may be difficult for the ANNs to pick up the complexity of phase transitions and their effect on viscosity. Perhaps the RE-modification is useful for viscosity predictions not considered in the case studies, or is better suited for pairing with a fully molten slag viscosity model not included in the SVP. The FactSage phase predictions used are another source of error.

Still, it may be worthwhile revisiting the assumptions made for the application of the RE-modification. The RE-modification relates the liquid phase viscosity ( $\eta_l$ ) and the solid-liquid mixture viscosity ( $\eta$ ) for a solution of dilute spheres or spheres with diverse sizes [27]:

$$\eta = \eta_l (1 + pf)^{-2.5}$$

where  $f$  is the volume fraction of solids,  $p$  is the solids interaction parameter. For a highly dilute concentration of spheres and/or with spheres of very diverse sizes, the solids do not interact with each other and the interaction parameter takes a value of 1. With higher concentrations of solids and/or with solids having similar sizes, the solids will immobilize liquid between them, increasing the apparent solid volume fraction in the mixture. In the case of equally-sized solid spheres, the maximum volume fraction which they can occupy in a given space is 0.74 [28]. In the extreme case of all the spheres aggregating for their densest packing and freezing all the liquid between them, the apparent solid volume fraction increases by a factor of 1.35 (i.e.,  $1/0.74$ ). Hence 1.35 should be used as the solids interaction parameter for high concentrations of equally sized solid spheres. With lower solid volume fractions of spheres which lead to the occasional freezing of liquid, the volume of frozen liquid as well as the frequency of interactions should be considered. For dilute equally-sized solid spheres, Roscoe [27] estimates that the solids spend a fraction of their time equal to  $4f$  as doublets. While forming doublets the spheres will freeze some liquid and double their effective volume. As a result, the solids interaction parameter is equal to  $1 + 4f$ . This neglects the formation of triplets and

higher order aggregates. It also assumes that the aggregates of solids with frozen liquid behave like spheres. The exponent of -2.5 in the equation originates from Einstein's equation for the viscosity of solutions with dilute rigid spheres [29,30]. For solutions with non-spherical particles, the exponent will take on different values [16]. Improvements to the RE-modification can include better approximations of the solids interaction parameter and of the exponent in the equation above. This will require greater knowledge of the sizes and shapes of solids in slags which can be determined experimentally and/or by fitting viscosity data to the viscosity model equations.

### **9.3 Modeling transformations of inorganic matter in entrained-flow slagging gasifiers**

Computational fluid dynamics (CFD) modelers face many obstacles when it comes to incorporating inorganic matter transformations in entrained-flow slagging gasifier simulations. The transformations are complex and include multiple phenomena; ash particle formation, gas-particle transport, particle sticking, slag flow and slag-refractory interaction (refer to Chapter 2). Other than some bench-scale experiments considering a subset of these, very little validation data is available for comprehensive gasifier simulations. In part, this is due to the harsh conditions in the gasifier and the difficulty in obtaining data relevant to the inorganic matter transformations. Furthermore, the gasifier data cannot be properly interpreted without thorough characterization of the fuels [31]. Many fuel properties are also required as inputs to the CFD model.

In Chapter 7, the characterization of four coals and a limestone is presented. Properties for fuel screening, gasifier operation, gasifier design and gasifier CFD modeling were evaluated. Fuel composition, inorganic matter composition and ash melting temperatures were applied for screening. Based on these properties it was determined that one of the coals is not a good fuel candidate and therefore would not be evaluated in subsequent testing. Slag viscosity was modeled and measured for the remaining fuels to determine

optimal fluxing ratios for gasifier operation. Slag-refractory reactivity was modeled and measured for the fuels or optimal fuel blends to assess the suitability of three refractory materials for gasifier design. Finally, the fuel forms and particle sizes, as well as the inorganic matter forms, particle sizes, associations, interfacial tension, density, heat capacity, heat conductivity and emissivity were discussed in terms of their application to CFD modeling. This work can serve as a blueprint for subsequent fuel characterization studies.

In Chapter 8, results from five pilot-scale gasification tests are presented. These tests utilized the fuels characterized in Chapter 7. For each test, solid samples were collected from the refractory liner, the gas probe sheath and impingers, the slag tap, the slag pot, quench discharge water and scrubber water. From these, ash and carbon mass balances were completed. A very small amount of solids was entrained with the syngas to the scrubber. Most of the solids were collected from the slag tap and slag pot. Carbon conversions ranged 97.0-99.6%. Most of the unconverted carbon was found in the slag pot. The majority of the inorganic matter melted in the gasifier and formed spheres due to surface energy minimization. The spherical inorganic matter on the char particles indicates that the liquid slag on the char is non-wetting. This confirms assumptions made in ash particle formation models [32,33]. Slag collected from the gasifier was affected by devolatilization and interaction with the gasifier refractory. This can significantly impact slag properties such as viscosity. Signs of spalling and/or erosion were detected. Slag layers on upper gasifier refractory liners were characterized in terms of thickness, composition, form and penetration into the refractory. In addition to validation data for CFD models, the results of Chapter 8 provide insight on often overlooked features of inorganic matter transformations in entrained-flow gasifiers.

One slag property which should be more thoroughly characterized is interfacial tension. The interfacial tension plays a critical role in particle sticking and slag-refractory interactions (refer to Chapter 2). Experimental data and models for slag surface tensions are available in the literature [6,34]. Data regarding slag-solid interfacial tension (and resulting contact angles) is sparser. It is particularly challenging to find data with the

right combination of slag and solid surfaces for specific applications. Data relevant to the slags and refractory materials used in the above-mentioned pilot plant study could help determine the mechanisms involved in particles sticking to the walls and slag penetration into the refractory. It could also help in the understanding of the spotting, rivulet formation and bubbling observed on the refractory liners.

As was mentioned in the introductions of Chapters 7 and 8, the fuel characterization and pilot plant testing are the first two parts of a three-part research program to develop an entrained-flow slagging gasification CFD model which includes inorganic matter transformations. The final part of the program involves creating the CFD model and using it for various simulations. The CFD modeling group at CanmetENERGY has already performed simulations for CanmetENERGY's pilot-scale gasifier [35,36]. However, these simulations did not consider the transformations of inorganic matter. It is important to include these transformations since they affect the overall gasifier performance, and proper slag flow and slag-refractory interactions are primary concerns for gasifier operators [37,38]. Sub-models discussed in Chapter 2 will be applied and flexibility to test different sub-models for each inorganic matter phenomenon is desired. Recommendations for the modeling of inorganic matter transformations include:

- Predict particle sticking via kinetic energy dissipation and/or critical char conversion.
- Consider particles settling on or being entrapped by the slag layer on the wall.
- Carefully select and apply slag viscosity models.
- Use cylindrical coordinates when modeling slag flow.
- Consider the effects of slag-refractory interactions on slag properties.

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## Appendix A

The rotational viscometer used at CanmetENERGY has a cylindrical crucible, a spindle with a cylindrical bob on its end, and a torque sensor. The sample is placed in the crucible and the bob is centrally immersed in the sample. The spindle is attached to the torque sensor. The torque required to turn the spindle at a known angular velocity is a function of the sample's viscosity.

A Brookfield RVDV-IIIU rheometer (full scale torque  $7.187 \times 10^{-4}$  N·m) is currently used for the CanmetENERGY system. The rheometer was calibrated with NIST's Standard Reference Material (SRM) 717a, a borosilicate glass. RheoCalc software is used to control all aspects of the rheometer directly from a computer. A Carbolite BLF 17/3 furnace is used to heat up the samples. Graphite sleeves are placed over sample crucibles to remove trace oxygen from the gas atmosphere. The sleeves are cylindrical with 56 mm OD, 50 mm ID and 103 mm length.

Molybdenum and alumina were chosen as the materials for the spindles and crucibles. Sizing and geometry of the first generation molybdenum spindles and crucibles are given in Figures A.1 and A.2, respectively.

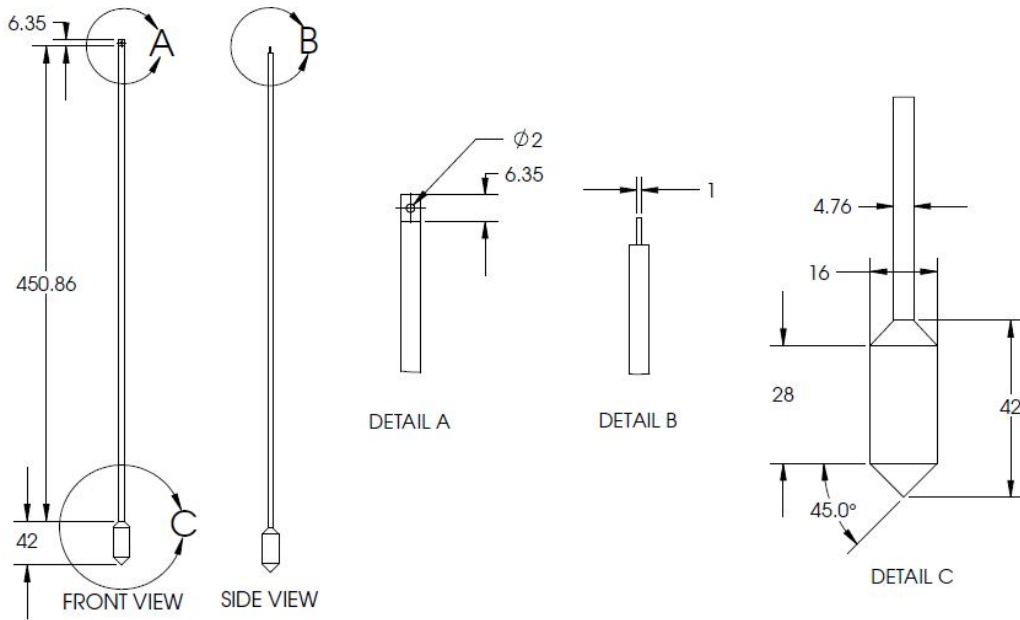


Figure A.1: First generation molybdenum spindle specifications. Dimensions are in millimetres.

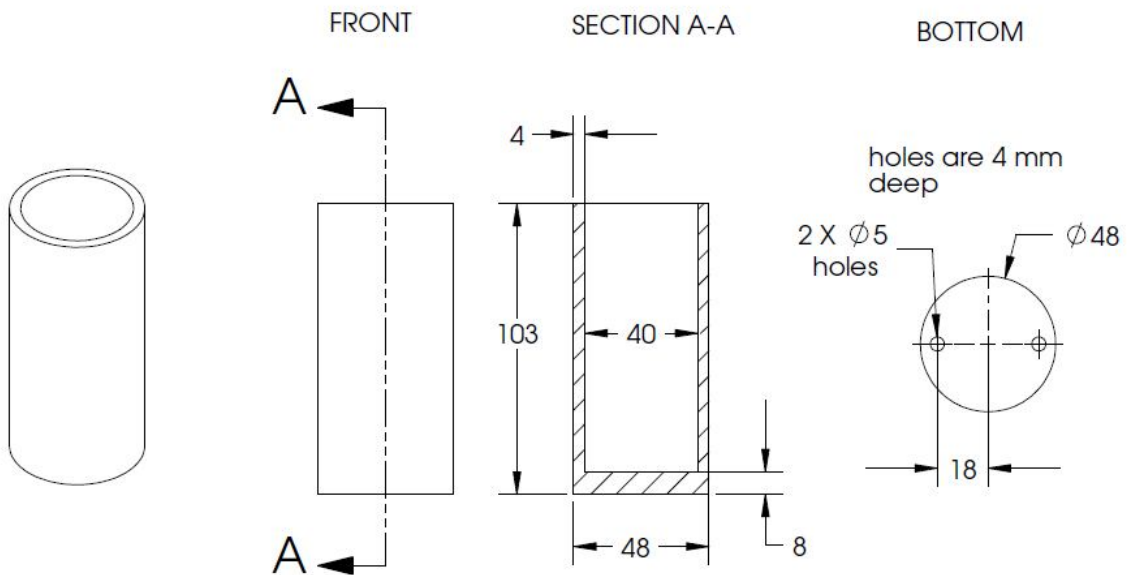


Figure A.2: First generation molybdenum crucible specifications. Dimensions are in millimetres.

Second generation molybdenum spindles (Figure A.3) were designed with a shorter bob in order to reduce the amount of sample required for measurements. The shorter bob also reduces the torque required for a given shear rate and viscosity. Second generation

molybdenum crucibles (Figure A.4) are taller to reduce the possibility of slag overflow. The walls are thinner to reduce cost.

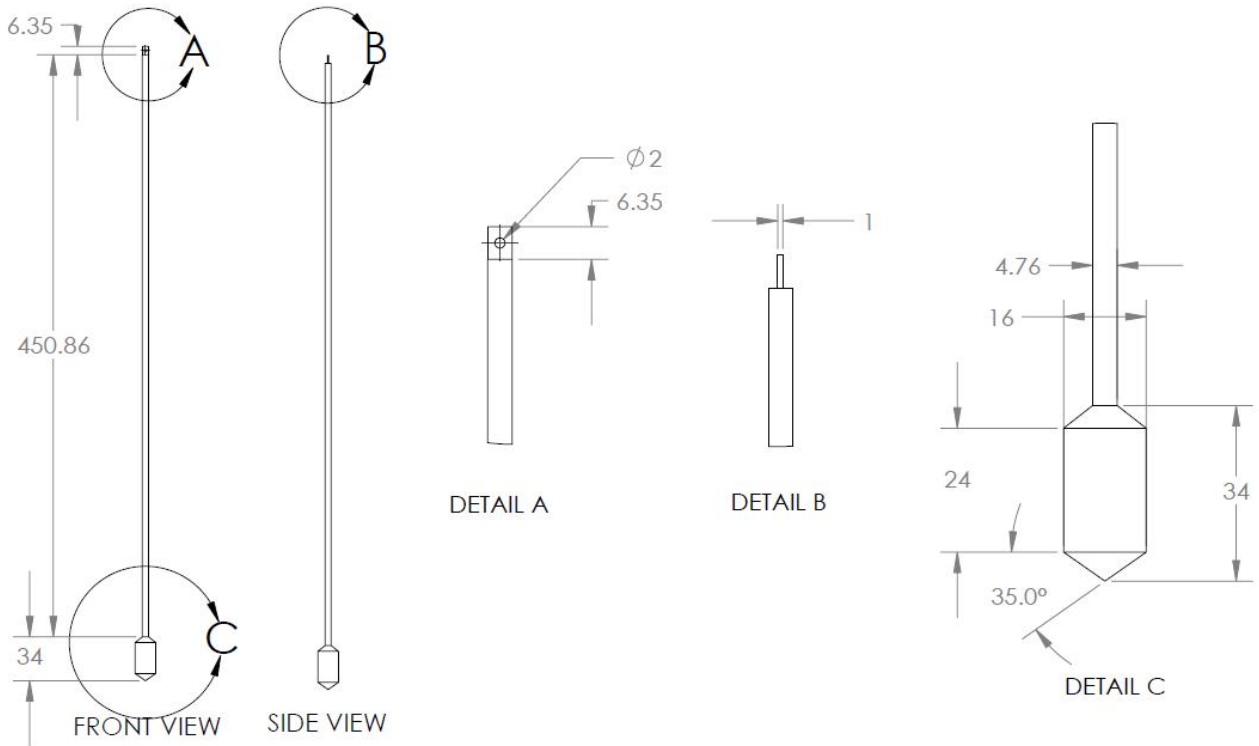


Figure A.3: Second generation molybdenum spindle specifications. Dimensions are in millimetres.

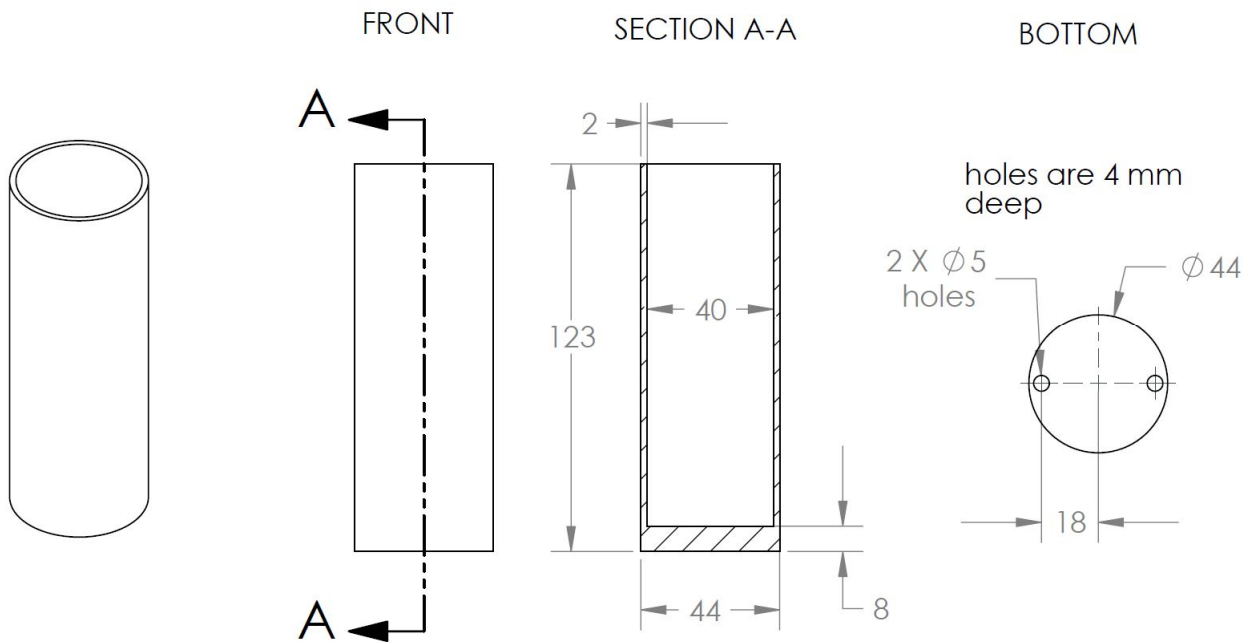
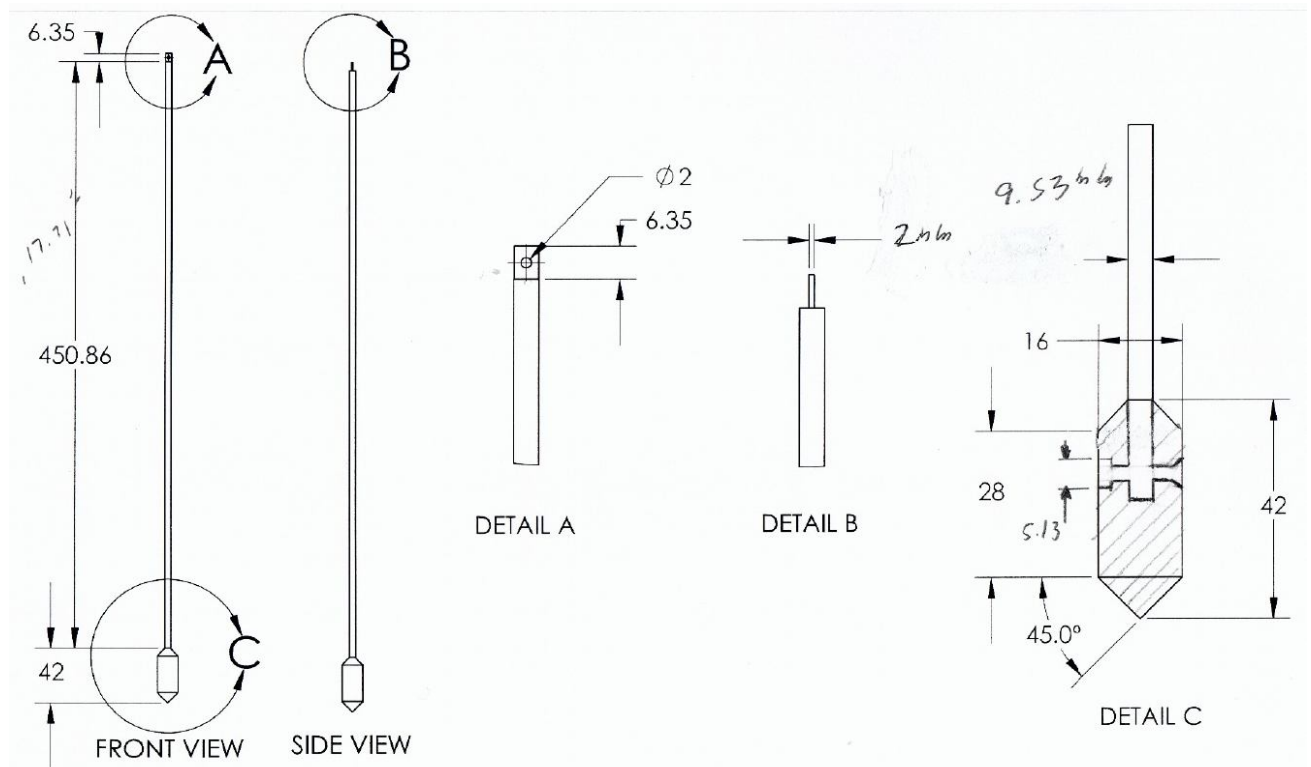
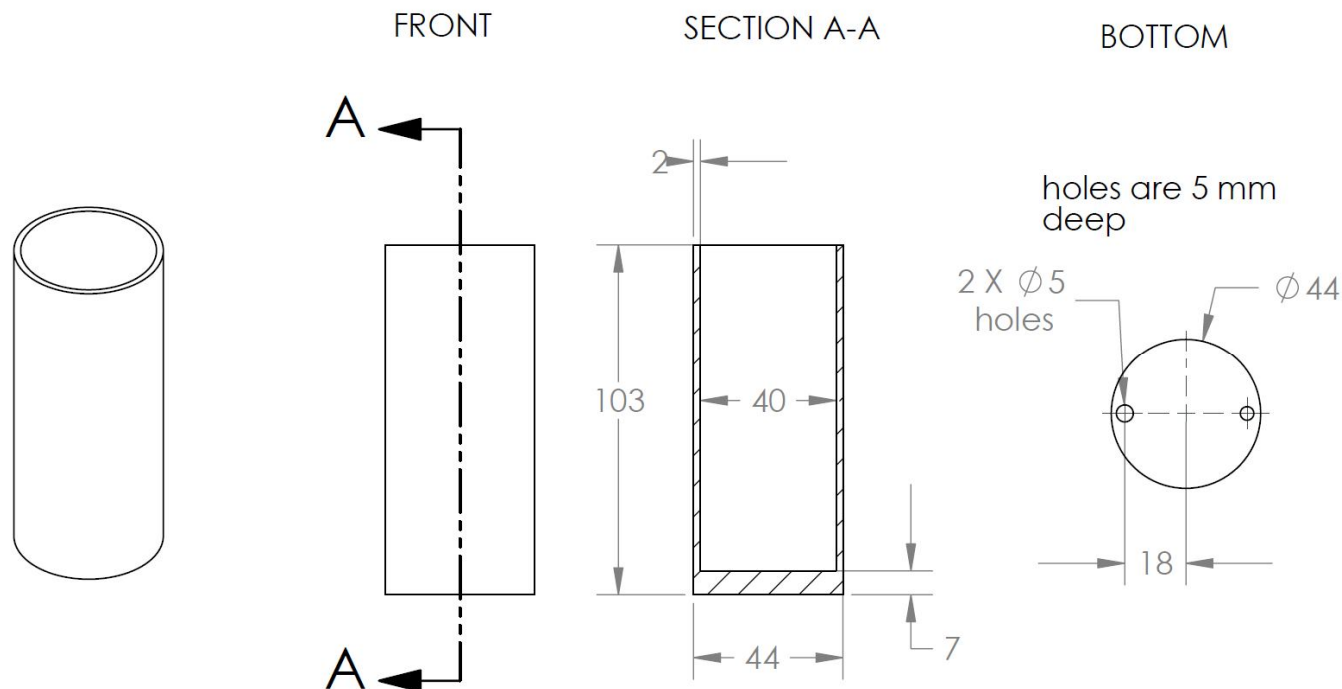


Figure A.4: Second generation molybdenum crucible specifications. Dimensions are in millimetres.

Alumina spindles (Figure A.5) and crucibles (Figure A.6) are used for testing with slag samples which react with molybdenum, or for viscosity tests in oxidizing conditions. The alumina spindle has an alumina bolt and nut which keep the shaft and bob attached.

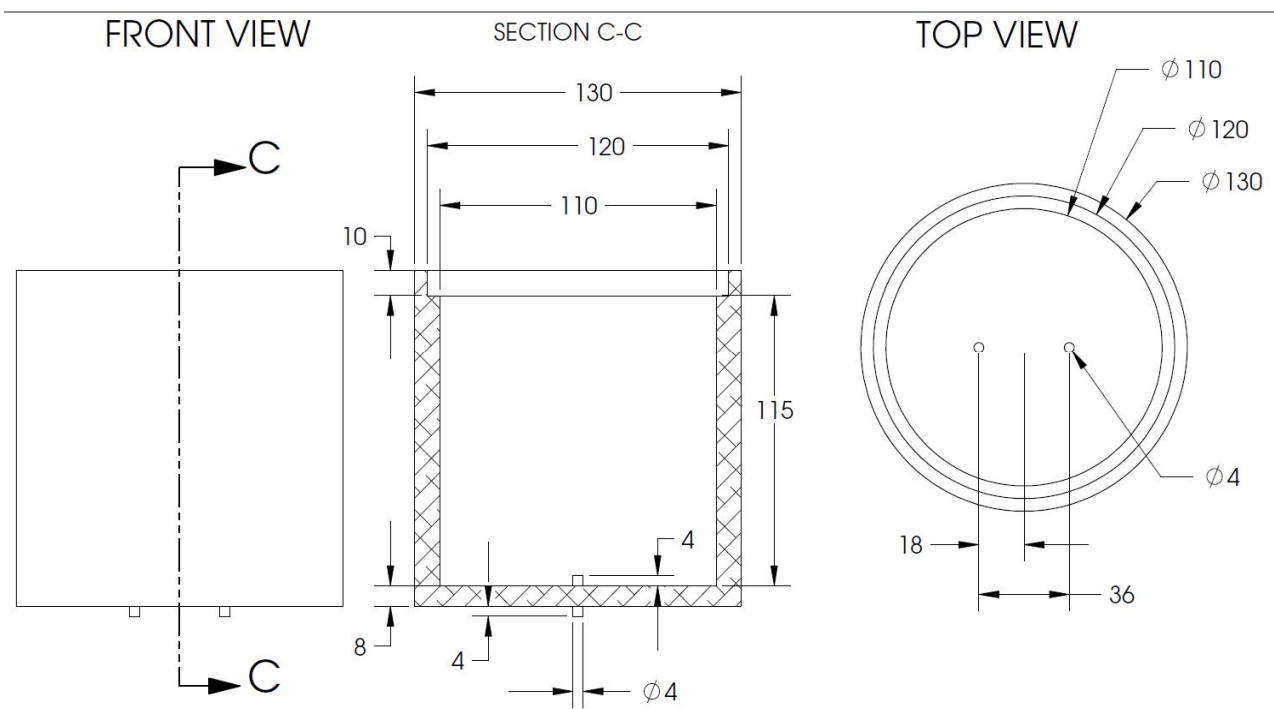


**Figure A.5: Alumina spindle specifications. Dimensions are in millimetres.**

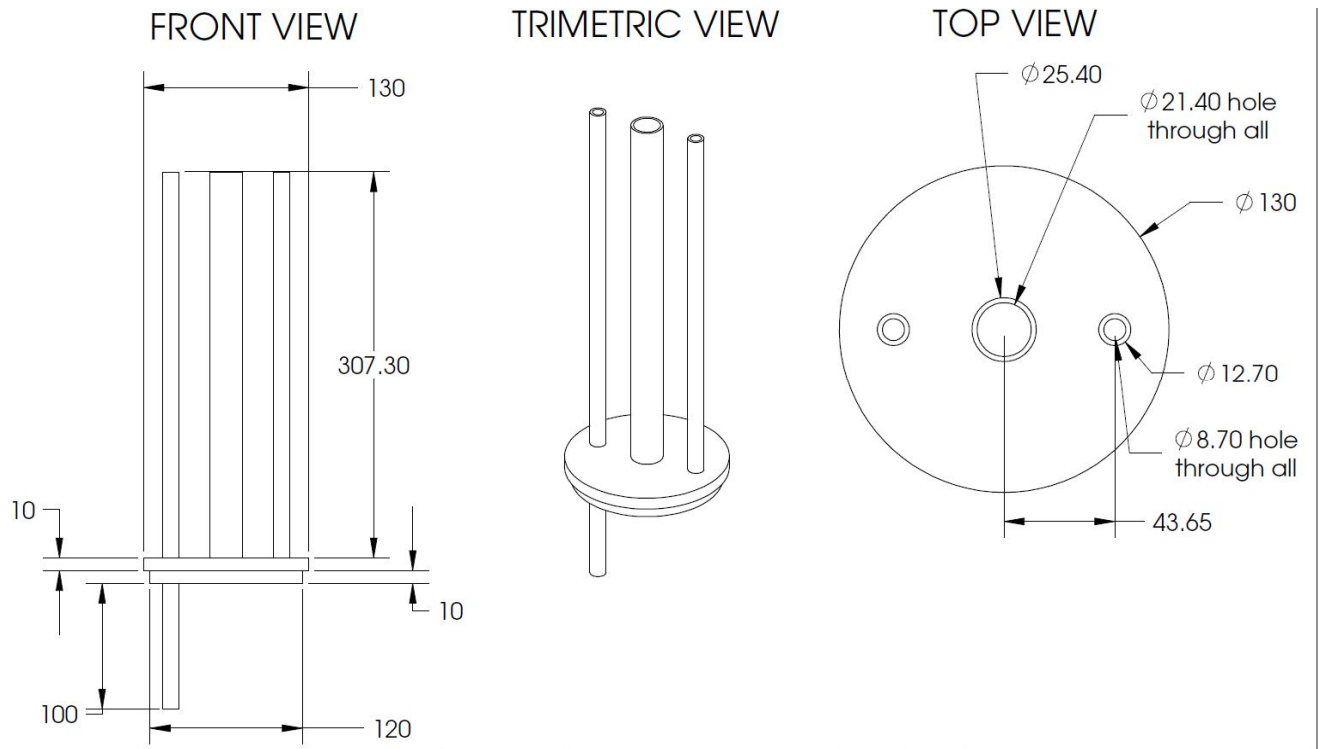


**Figure A.6: Alumina crucible specifications. Dimensions are in millimetres.**

Since the furnace is not gas tight, a custom alumina chamber to contain the crucible and sample is used. This chamber also protects the furnace in case the slag overflows, splashes or goes through the sample crucible. The alumina chamber consists of two parts, a base piece and a cap piece. The design of the chamber incurred multiple iterations. The first generation base and cap pieces are shown in Figures A.7 and A.8, respectively. The cap sits on top of the base to form the chamber. There were multiple problems with the first generation design. The base would crack upon heating. This was caused by the temperature gradient formed in the bottom which sits on furnace insulation. To solve this problem, the base is now placed on small alumina posts which allow better heating of the base bottom. Another problem with the first generation design is that the cap piece can stay stuck in the base piece upon a heating and cooling cycle. Hence the diameter of the rabbet on the cap piece was reduced by grinding. The purpose of the rabbet fit between the cap and base is to provide a better gas seal, and to ensure the base is centered with the cap. However, it was determined with experimentation that the rabbet is not necessary. The pins in the bottom of the base anchor the sample crucible from rotation. The sample crucible sits on small posts placed on the bottom of the base.

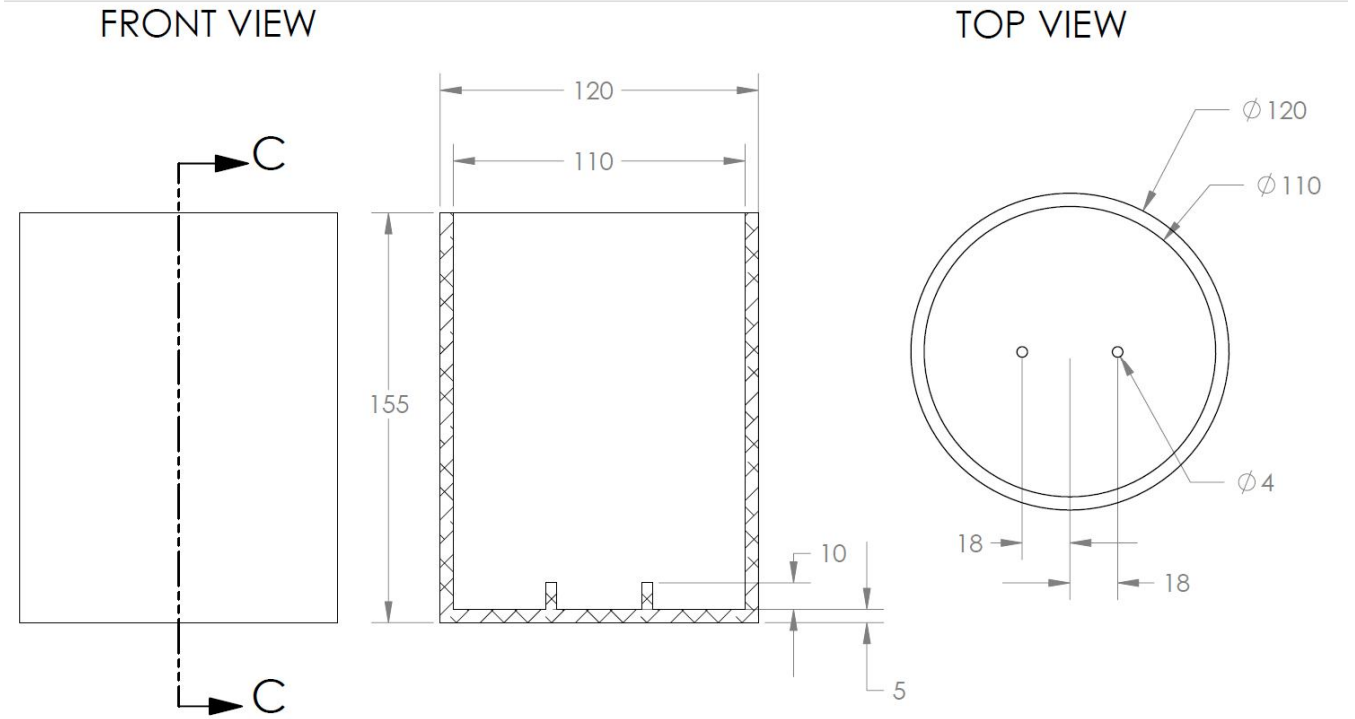


**Figure A.7: First generation base specifications. Dimensions are in millimetres.**

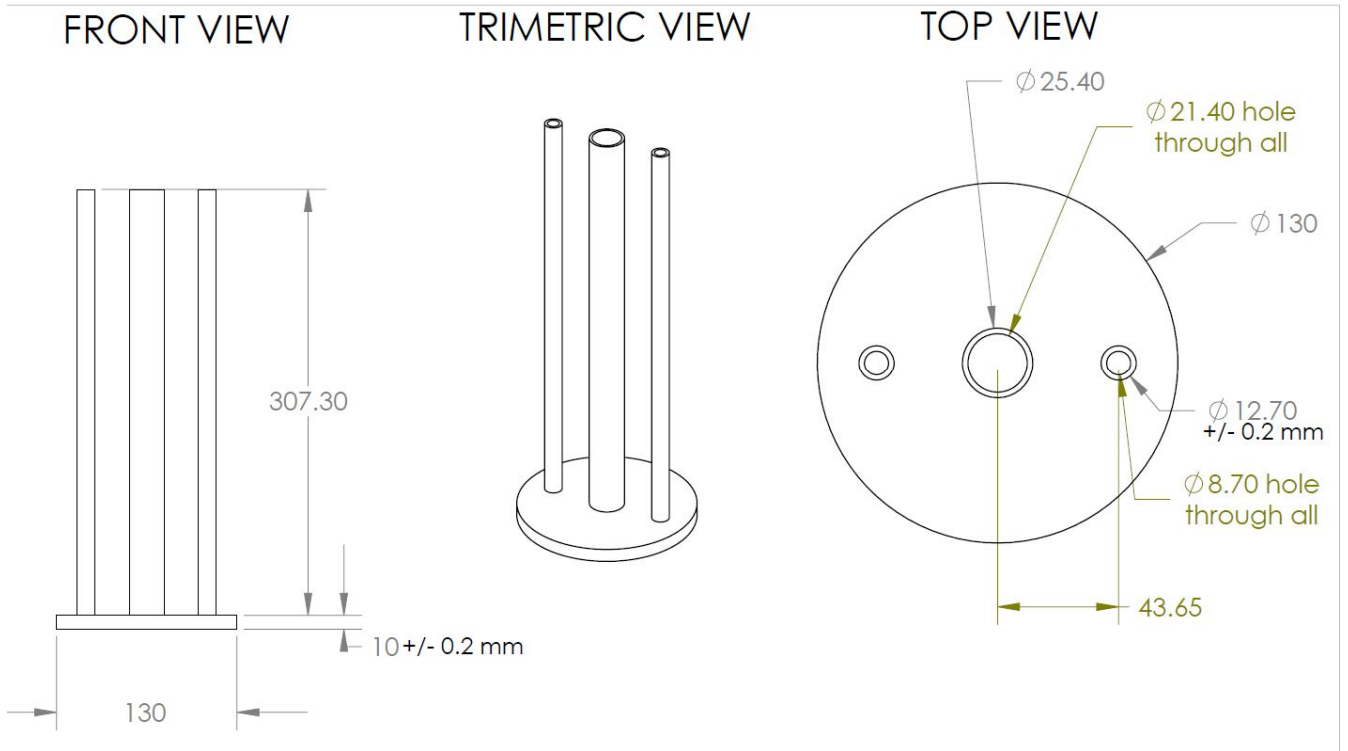


**Figure A.8: First generation cap specifications. Dimensions are in millimetres.**

The second generation base and cap pieces are shown in Figures A.9 and A.10, respectively. The second generation base is thinner to increase heat transfer and to make it easier to handle. It is taller to permit the use of the taller second generation molybdenum sample crucibles. There is no longer a rabbet between the cap and base. Also, the cap no longer has a bottom tube.



**Figure A.9: Second generation base specifications. Dimensions are in millimetres.**



**Figure A.10: Second generation cap specifications. Dimensions are in millimetres.**