Use of H₂ in Mn-ferroalloy production

Merete Tangstad¹, Trygve Schanche¹, Faan de Preez²

- 1. Norwegian University of Science and Technology, Trondheim, Norway
- 2. North West University, Potchefstroom, South Africa

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Abstract

The use of H_2 as a reductant in the iron and steel industry is an obvious choice towards carbon neutrality. For pyrometallurgical processes like Mn and Si, H_2 cannot be the only solution, as the stability of the oxides of these elements are higher compared to Fe. H_2 can however be used in the Mn-process together with other low CO_2 emission mitigations. In the Mn-ferroalloy process H_2 can be used to reduce higher manganese oxides to MnO, and the last part of the reduction to metallic Mn can be done with biocarbon or with electrolysis. In studies from NTNU, the reduction of higher manganese oxides to MnO has been investigated with pure H_2 or with CO/H_2 mixtures. It is shown that the H_2 containing gases will give higher reduction rate compared to CO/CO_2 gases with the same reduction potential. The reduction rate will be increased in the order of 20-100%, and H_2 has a larger impact on reduction rate for high oxygen pressures. In H_2 gas it is seen that the final degree of reduction is higher than using CO gas, and this is believed to be due to the formation of metallic iron. The reduction rate in H_2 gases varies for various ores, and as for CO containing gases, the reduction rate of Comilog ore is faster than the Nchwaning ore, and activation energies of 23 kJ/mol versus 68 kJ/mol is respectively found for isothermal experiments in CO, H_2 , CO_2 gases.

1 INTRODUCTION

Mn ferroalloys are mainly produced in Submerged Arc Furnaces (SAF) as many other ferroalloys. The most typical geometry is the circular furnace with 3 Søderberg electrodes. In the lower part of the furnace the electrical energy is provided by the electrodes, and here the temperature is 1400-1600 °C. At these temperatures the oxide materials are melted and reduced to metal, before it is drained from the furnace as metal and slag¹. The slag/metal ratio is typically in the area of 0.5-1. The raw material mix, containing Mn-sources, fluxes and carbon materials, are being fed to the charge by gravity at the top of the furnace, where the temperature is at its lowest, in the area of 200-600 °C. As slag and metal are being tapped from the furnace the rest of the material in the furnace will descend while heated by the gas coming from the high temperature zone. The high temperature zone is called the coke-bed (1400-1600 °C), due to a stagnant layer of carbon materials, while the low temperature zone (200-1400 °C) is called the pre-reduction zone.

In the high temperature zone, the following carbon consuming reactions occur in the metal producing reactions during High Carbon FeMn (HC FeMn) production²:

MnO+C=Mn+CO	(1)
FeO+C=Fe+CO	(2)
$C=\underline{C}$ (carbon dissolves into the metal)	(3)

Analysing the CO_2 emission, one can start to discuss the total carbon consumption per ton of metal produced. The typical composition of HC FeMn is 76-79%Mn, 7%C and the rest being iron. As shown in Table 1, this requires 273 kg carbon, however only 203 kg will go the gas phase as CO, and the rest will dissolve in the metal. There is however also some carbon consumption in the prereduction zone. The higher Mn- and Fe-oxides is reduced with CO gas to CO_2 (Rx.4) and if this reaction happens above about 800 °C, the CO_2 will subsequently react with the carbon materials according to the Boudouard reaction (Rx.5), and the sum of these two reactions (Rx.6) will consume carbon^{2,3}.

$Me_yO_x+CO=MeO_{x-1}+CO_2$	(4)
$CO_2 + C = 2CO$	(5)
$Me_yO_x+C=Me_yO_{x-1}+CO$	(6)

 CO_2 from the decomposition of carbonates may also react according to the Boudouard reaction (Rx.5), and hence the total reaction will be according to Rx.7:

$CaCO_3 + C = CaO + 2CO \tag{7}$

Based on industrial operation over 5 years and 4 different furnaces³, it has been seen that the CO₂ from the reduction of MnO_2 , Mn_2O_3 and Fe_2O_3 with CO gas, will not react with solid carbon, and hence the total oxygen content of the charge mix will not affect the total carbon consumption. The CO₂ from Mn_3O_4 and Fe_3O_4 to MnO and FeO, as well as the CO₂ from the decomposition of lime (CaCO₃), may or may not react with carbon to CO gas. If none of this CO₂ is reacting with C, it has been defined as 100% degree of prereduction, and if all of this CO₂ reacts C it has been defined as 0% degree of prereduction. Industrially, it was seen that the typical degree of prereduction was changing between 0% to 60%. The degree of prereduction is determined by type and size of Mn-sources as well as furnace operation.

So far we have discussed the carbon consumption, as it is proportional with the final CO₂ emissions (with the exemption of the carbon dissolved in the metal). This is always the case for open furnaces where all CO gas will burn on the charge top. There are however some notes that must be included on this issue. For closed furnaces, the off gas will be a mixture of CO and CO₂ gas. If the gas is flared, again the carbon consumption will be proportional with the C consumption. As the off gas from Mn-alloy production may contain high amounts of CO (50-90%), this gas may be sold to other users of CO as e.g., fuel. When this gas is burned to CO₂ and emitted, the total amount of CO₂ emissions will again be proportional with the original carbon consumption. This CO₂ may however be reported as a part of the customers CO₂ emissions and not the metal producer. From an environmental point of view, the total CO₂ emission will still be proportional with the carbon consumption in the metal producing furnace, and the goal is to reduce this to a minimum.

In Table 1 the carbon consumption needed in the HC FeMn is shown. In the high temperature zone 273kg of carbon per ton of metal is required, however 70kg of these will be dissolved in the metal and hence 203 kg C or 743kg CO_2 will be emitted from the reactions in this zone. If no prereduction is occurring, the additional carbon consumption in the prereduction zone is 81 kg C, that is 341kg CO_2 , will be emitted. This means that with these operating conditions, 31% of the CO_2 emitted from the furnace will come from the prereduction zone.

		kg C	$kg CO_2$
790 kg Mn	MnO+C=Mn+CO	173	633
140 kg Fe	FeO+C=Fe+CO	30	110
70kg C	C dissolved in metal	70	0
C in cokebed zone		273	743
s/m=0,7, 40%MnO	Mn ₃ O ₄ +C=3MnO+CO	59	217
	Fe ₃ O ₄ +C=3FeO+CO	10	37
100kg CaCO3/ton metal	CaCO ₃ +C=CaO+2CO	12	88
0% degree of prereduction		81	341
Total		354	1084

Table 1. Example of carbon consumption and subsequently CO_2 emission producing a metal with 79%Mn, a slag/metal ratio of 0.7, 40% MnO in the slag, 100kg lime/ton metal and 0% prereduction.

One of the strategies to reduce the CO_2 emissions in today's processes is to increase the degree of prereduction in the furnace, that is to reduce the extent of the Boudouard reaction, and hence reduce the total carbon and energy consumption. Figure 1 shows the carbon consumption and the CO_2 emission for the same charge as shown in Table 1 as a function of degree of prereduction. As previously noted, the highest degree of prereduction seen industrially was in the area of 60%, and this would reduce the CO_2 emissions to 870 kg CO_2 , that is 19% reduction. To obtain such numbers, the operation is optimal under today's conditions. The main factor for this to happen is first and foremost, good raw materials. If the raw materials could be developed further to obtain 100% degree of prereduction, a total reduction of 31% CO_2 emission compared to 0% prereduction could be obtained.



Figure 1. C consumption and subsequently CO_2 emissions for a given charge (slag/metal ratio=0.7, %MnO in slag=40%, 100kg CaCO₃ per ton metal, all Mn-sources are MnO₂, Mn₂O₃ or Mn₃O₄).

One of the discussed paths to reduce CO_2 emissions is to use H_2 instead of solid carbon. MnO is a stable oxide and cannot be reduced with H_2 , only with C as shown in reaction 1. FeO and the higher manganese oxides can however be reduced with H_2 according to reaction 8 and 9 as seen in Figure 2.



Figure 2. Ellingham diagram of Mn- and iron-oxides compared to H_2/H_2O and CO/CO_2 gases (calculated from HSC Chemistry10).

 $Me_yO_x+H_2=Me_yO_{x-1}+H_2O$

FeO+H2=Fe+H2O

In a process where the reactions of the prereduction zone could be separated from the high temperature cokebed zone, e.g. in a prereduction unit, H_2 could be used to reduce the higher manganese oxides and the FeO without any Boudouard reaction. A more possible scenario is to operate a prereduction unit with no solid carbon, and hence both the CO from the furnace and/or additional H_2 may be used. In both cases, the total emitted CO₂ would now be down to 633 kg CO₂ per ton of metal produced (or 743 kg CO₂ if the iron is reduced to metallic iron in the prereduction unit). The conditions, which is required to reduce the ore down to MnO and Fe is investigated in a number of projects^{4–8}. This paper will sum up the work done at the Norwegian University of Science and Technology within the use of H_2 as a reductant for Mn-ores.

2.1. Methods and materials

Nchwaning and Comilog ore was the raw materials used in most of the studies, in addition to one study using UMK ore. UMK is a semi-oxidized (Mn_2O_3) carbonate ore. The main minerals are braunite I $(3(Mn,Fe)_2O_3.MnSiO_3)$, carbonates (calcite (CaCO₃), dolomite (CaMg(CO₃)₂), and hausmannite (Mn₃O₄). Kutnahorite ((Ca,Mn,Mg,Fe)(CO₃)₂), bixbyite ((Mn,Fe)₂O₃), and hematite (Fe₂O₃) is also found but to a lesser degree.^[14] Comilog is a high oxygen ore, where the oxygen level is close to that of MnO₂. The ore is low in iron and contains no carbonates. It contains approximately 5 wt% chemically bound moisture found in minerals such as lithiophorite and nsutite, as recently reported^{5,9,10}. Nchwaning ore is a semi-oxygen ore, where the oxygen level is close to 1.5 correlating to Mn₂O₃. It contains insignificant amounts of chemically bound moisture but has a relatively high iron-content and smaller amounts of carbonate. The chemical analyses for the ores is given Table 2.

A TGA technique was used to investigate prereduction. The experiments were performed in an Entech VTF 80/15 vertical resistance tube furnace. The ore was placed in a high-temperature gas-tight stainless-steel (steel grade 253 MA) double-wall crucible (height 45 cm and diameter 4.8 cm). The double wall crucible ensures preheating and premixing of the gas. The gas outlet is connected to an off-gas analyser (NDIR) which determines the CO and CO₂ concentrations in the off-gas. The crucible was suspended from a balance (Mettler Toledo PR2003DR, Switzerland, 10mg) to record the weight changes during the experiment. The heating of the furnace was controlled by a Eurotherm PID controller and measured with a calibrated S-type thermocouple. A K-type thermocouple placed in an alumina tube and positioned in the middle of the charge was used to measure the charge temperature.

Flexible gas inlet and outlet pipes were connected to the top of the crucible which allowed multiple gasses to mix before entering the crucible. Each gas had its own mass flow controller (Bronkhorst F-201C). The weight of the sample before and after the experiment was recorded to confirm the net weight loss obtained from the TGA data. After the experiment was complete the sample was allowed to cool in Ar.

The chemical composition of the raw materials and the prereduced ores was examined by X-ray fluorescence (Bruker AXS S4 Pioneer X-Ray fluorescence spectrometer) using the fused bead technique. Permanganometric titration (ASTM 465-11:2017) was employed to measure the excess oxygen above MnO which is expressed as MnO_2 . The carbon concentration was analysed with a LECO (Combustion-IR) instrument and the CO_2 concentration was determined by assuming that all carbon in the ore is in the form of CO_2 .

	Fe, tot	Mn, tot	MnO ₂	MnO	CaO	SiO ₂	Al ₂ O ₃	K ₂ O	CO ₂
	[wt%]	[wt%]	[wt%]	[wt%]	[wt%]	[wt%]	[wt%]	[wt%]	[wt%]
Comilog ⁶	3.1	51.0	76.4	3.5	0.1	3.5	5.6	0.7	0.1
Comilog ¹⁰	3.6	51.4	74.3		0.1	3.3	5.1	0.8	0.15
Nchwaning	10.0	46.4	34.6	31.6	5.9	6.7	0.5	0.0	3.0
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Ncwaning ⁸	8.7±2.5	48.7±3.3	38.8±2.0		5.9±0.6	4.4±0.1	0.4±0.1	0.0	2.5±0.5
UMK 11	5.9	33.7	20.55		17.0	7.3	0.4	0.06	17.0

Table 2: Chemical composition of ores (average±st.dev.)



Figure 3. Sketch of the crucible used in the reduction experiments⁶.

Table 3. Experimental	conditions of	the studies	discussed	in this article.
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	Lobo	Ngoy	Larssen	Davies	Schanche
Ores	Nchwaning	Nchwaning Comilog	Nchwaning Comilog	UMK	Nchwaning Comilog
	pellets	lumps	lumps	lumps	lumps
Size	10-14mm	9.5-16mm	11.2-15mm	9.5-16mm	9.2-15mm
Gas	various mixes of H ₂ ,H ₂ O,CO,CO ₂	CO/CO ₂ , CO/CO ₂ ,H ₂	CO/CO ₂ , CO/CO ₂ /H ₂ , CO/CO ₂ /H ₂ O	CO/CO ₂ , H ₂ /H ₂ O, H ₂	CO/CO ₂ , CO/CO ₂ /H ₂
	Isothermal	Non-isothermal	Non-isothermal	Isothermal*	Isothermal*
Gas flow	51/min	2.85, 4.851/min	41/min	4 l/min	41/min
Isothermal temp.	950 °C			700, 800, 900°C	605,700,790 °C (Nch) 365-605 °C (Com)

* Parts of the reduction will occur before isothermal temperature is reached

As seen in Table 3, the temperature profile and gas compositions vary in the different studies. For the isothermal experiments two strategies were chosen. Lobo heated the Nchwaning pellets in Ar to the isothermal temperature. Lobo also roasted the pellets for 2 hours in 800 °C to get a good mechanical strength. It has however been seen that the heating conditions is affecting the subsequent prereduction rate¹³, and Schanche and Davies hence chose to heat the ore in the gas species used during the isothermal reduction, to simulate an industrial process. The disadvantage for this method is that some of the reduction will occur during the heating. Also in the non-isothermal experiments, where a fixed heating rate was used, two different strategies was chosen. Larssen chose to use the same gas atmosphere during the whole temperature range. It was however seen that CO would decompose to C and CO₂ at lower temperature in H₂ rich atmospheres, and hence Ngoy heated the ore to 500 °C in CO₂, before adding the reducing gas composition from 500 to 1000 °C. As seen by both Ngoy¹² and Biørnstad¹⁴, Comilog ore would then decompose from MnO to Mn₂O₃ during the heating step, while Nchwaning would hardly change the oxidation degree.

2 RESULTS AND DISCUSSION

2.1 Reduction of Mn-ores in H2-H2O gases compared to CO-CO2 gases

Figure 2 shows the reducing potential of H_2/H_2O versus CO/CO_2 gases. For the same H_2/H_2O ratio as CO/CO_2 ratio, the CO/CO_2 gas will have lower O_2 -partial pressure below 800 °C and H_2/H_2O gas will have the lowest O_2 -

partial pressure, and thus the highest reduction potential, above this temperature. It can also be seen that the difference in reduction potential is not very large above 800°C; at 900 °C the O₂ partial pressure for a 70/30 H₂/H₂O gas is $5.5 \cdot 10^{-5}$ bar and for a 70/30 CO/CO₂ gas it is $6.3 \cdot 10^{-5}$ bar. The overall picture is hence that the CO and H₂ has quite similar reduction potential in the temperature range 500-1100 °C.



Figure 4. Comparison of reduction of lumpy material and pellets from Nchwaning ore with H_2 and 95%CO gas at 950 °C from Lobos study.



Figure 5. Extent of reduction and decomposition of carbonates in UMK ore at 700, 800 and 900 °C investigated by Davies¹¹.(Note a negative extent below 200 °C due to H_2O condensation from the H_2 - H_2O gas.)

Figure 4 shows the difference between the reduction rate and the extent of the reduction in CO and the H₂ system for Nchwaning ore and Nchwaning pellets. Though the reduction potential is quite similar as discussed above, it is seen that the H₂ gas is reducing the ore to a higher extent than the CO gas. If however a mixture of H₂ and CO is used, the extent of the reduction will be in the same area as the CO gas. The higher extent of reduction is also seen by Davies¹¹ as shown in Figure 4 when reducing the UMK carbonate ore. At 700 and 800 °C in CO-CO₂ gas, the carbonates are not yet decomposed as seen in Figure 6 which explains the lower extent of mass loss at these temperatures. At 900 °C, where all the carbonates are decomposed, the extent of reduction is higher in H₂ compared to H₂/H₂O and CO/CO₂ mixtures, which can be explained by the increased reduction potential in the H₂ gas as shown in Figure 7. H₂ gas will at equilibrium condition reduce some iron to metallic state, while the Mn will be in the (Mn,Fe)O phase. The 70/30 H₂/H₂O and CO/CO₂ gas will be at the boundary line between the metallic Fe and FeO line. According to the XRD as shown in Figure 8, it verifies that metallic Fe is produced in H₂ gas. In the mixed gases the metallic phase was not found. The difference in α when all FeO is reduced to Fe is 7% and hence it seems like most of the FeO is reduced to metallic state. Going back to the study of Lobo in Figure 4 again, the difference between H₂ and CO cannot be explained by the difference in reduction potential, as the equilibrium state at this temperature and gas composition will now be metallic for both gas mixtures. Metallic iron was found by XRD for both gas mixtures, however for the H₂ gas experiments the monoxide was MnO and for the CO gas it was (Mn,Fe)O. This also indicate that it is the difference in the iron reduction that separates the extent of reduction.



Figure 6. Fraction of the total wt.% of carbonates decomposed in UMK ore as a function of temperature.¹¹



Figure 7. The Fe-O and Mn-O stability diagrams show the stable phases with varying oxygen partial pressure and temperature calculated by Factsage and HSC chemistry. Left show the superimposed Fe-O and Mn-O diagram, and right shows solid solution Fe-Mn-O diagram^{10,11}.



Figure 8. XRD results of the ore and pre-reduced in 70% CO 30% CO₂ (a), 70% H₂ 30% H₂O (b), and 100% H₂ (c) at 700, 800, and 900 °C. Chemical formula of compounds: calcite (CaCO₃); magnesium carbonate (MgCO₃) braunite (3(Mn,Fe)₂O₃·MnSiO₃); bixbyite ((Mn,Fe)₂O₃); hematite (Fe₂O₃); kutnahorite ((Ca,Mn,Mg,Fe)(CO₃)₂); manganosite (MnO); calcium oxide (CaO); hausmannite (Mn₃O₄).¹¹

2.2 Reduction of Mn-ores in H₂-H₂O-CO-CO₂ gases

Ngoy ¹² investigated CO-CO₂-H₂ mixtures and compared it with CO-CO₂ mixtures with the same partial pressure of O₂ when in equilibrium, that is the same reduction potential. The samples were heated to 500 °C in CO₂ before the reducing gas was added. Both Comilog and Nchwaning ore was investigated with two oxygen partial pressures. The mass loss is shown in Figure 9. It is seen that the rate is higher in the H₂ containing gases in the low O₂ containing gases, that is Figure 9 a) and c). At higher oxygen pressures, Figure 9 b) and c) the H₂ containing gas reduces the Comilog ore faster. For the Nchwaning ore, the H₂ gas starts the reduction a bit later, however the rate is faster when it has started. This experiment may be an outlier as it does not fit the results in the later work done. In the H₂-containing gas experiments, carbon deposition was seen both visually after the experiment, and



on the weight curve, as shown by the increasing weight for Comilog ore. If carbon deposition occurred during the whole reduction path, it means that the H_2 containing gases will reduce the ore faster than the graph shows.

Figure 9. Comparison of mass loss for H_2 containing and H_2 free gas for Comilog (a and b) and Nchwaning ores (c and d). a9 and c) have the same low oxygen pressure and b) and d) has a higher oxygen pressure in the gas. (Note: In c) the two H_2 containing gas experiments have been shifted in time, to start the reduction at the same time as the H_2 free gases in c).)¹²

Larssen⁶ investigated the prereduction of Comilog ore with increasing temperature for two oxygen partial pressures, with and without H₂, as seen in Figure 10. Both for the high oxygen partial pressures experiments, the 50/50% CO/CO₂ and 44/12/44% CO/H₂/CO₂ experiments, and the low oxygen partial pressure experiments, the 80/20% CO/CO₂ and 70/12/18% CO/H₂/CO₂ experiments, the hydrogen has a strong effect on the reduction rate. It was also noted that the C content after the experiments was above1% for the H₂ containing gas and about 0.2-0.3 in the CO-CO₂ experiments, which again means that carbon was deposited during the experiments, and hence that the real weight loss from the reduction was even higher. In the low oxygen partial pressure experiment, one did not experience the typical temperature peak from high oxygen ores just below 600 °C, where the remaining MnO₂ would decompose to Mn₂O₃⁵. This is explained by the fast reduction due to the low oxygen partial pressure and H₂ being present.

Similar experiments were performed using Nchwaning ore, however they were influenced by carbon deposition⁶. For both oxygen partial pressures, the carbon deposition was initiated above 400 °C. This correlates well with previous reports where carbon deposition is initiated at 400°C for this type of system, and the deposition extent is promoted by an increased CO-concentration in the CO-CO₂ atmosphere⁹. The carbon deposition also increased with increased H₂ pressure as seen by Ngoy et al. ¹². Compared to Nchwaning ore, Comilog ore produce more CO₂ during the reduction with CO gas, and carbon deposition occur to a lesser extent.

Schanche 8,10 investigated the isothermal prereduction for Nchwaning and Comilog ore with and without H₂ present in CO/CO₂ gases as seen in Figure 11 and



Figure 12. As for previous studies the H_2 added to the CO/CO₂ gases increased the reduction rate, mainly in the area of 30-50%, with the exemption of high oxygen gas mixture and Nchwaning ore, where the H_2 in the gas mixture more than doubled the rate. Ngoy also found in his study that the H_2 in the gas increased the rate with 20-40%.



Figure 10. Weight loss for non-isothermal reduction of Comilog with H_2 containing and H_2 -free gas composition, where the grey lines have high oxygen pressure, and the blue lines have low oxygen pressure⁶.



Figure 11. The effect of hydrogen at constant oxygen partial pressure at 790 °C for (a) low pO_2 , and (b) high pO_2 (Nchwaning ore)¹⁰.





Figure 12 The fractional conversion curves for Comilog ore; (a) - hydrogen free gas mixtures at 410 °C, (b) - hydrogen containing gas mixtures at 410 °C, (c) - low pO_2 at 365 °C, (d) – medium pO_2 at 410 °C, (e) - high pO_2 at 455 °C¹⁰.

The reduction rate can be assumed to be a function of the reduction potential, temperature and the extent of the reaction as well as the presence of H_2 in the reduction gas. Based on this, Schanche ¹⁰ modelled the reaction rate according to the following equation.

$$\frac{d\alpha}{dt} = k(T)f(\alpha) = k_0 \exp\left(-\frac{E_a}{RT(t)}\right)(1-\alpha)^x.$$
(10)

To find the dependency of the gas composition the logarithm of k_0 is plotted as a function of $ln(1-CO_2)$. The parameters m and k are determined based on the slope and y-intercept of the linear regression between the points as shown in



Figure 13. Based on the correlation, the total equation describing the reaction rate is given in equation (10).



Figure 13: The dependency of reaction rate on gas composition for (a): Nchwaning ore and (b): Comilog ore ¹⁰.

$$\frac{d\alpha}{dt} = k \exp\left(-\frac{E_a}{RT(t)}\right) (1-\alpha)^x p^m_{(1-CO_2)} \tag{11}$$

Table 4 contains the kinetic parameters in Schanches work¹⁰. The determined kinetic parameters can be used for reduction of Nchwaning and Comilog ore in CO/CO_2 gas mixtures and in CO/CO_2 gas mixtures containing hydrogen where the H₂/CO ratio is 1.

parenthesis are excluding the medium pO_2^{10}							
Parameter	Nchw	aning	Co	milog			
	CO/CO_2	$CO/CO_2/H_2$	CO/CO_2	$CO/CO_2/H_2$			

Table 4. Values obtained from modelling of Nchwaning and Comilog ore using equation 11. Numbers in

		CO/CO_2	$CO/CO_2/H_2$	CO/CO_2	$CO/CO_2/H_2$
k	Rate constant	75.9	69.4	3.82 (5.05)	6.18
Ea	Activation energy	68.1	68.1	23.2	23.2
x	Reaction order	2	2	2	2
т	Order of 1-CO ₂	1.21	0.38	0.72 (0.82)	0.65

There are a couple of issues when doing experiments with $CO-H_2-CO_2$ gases assuming the same O_2 partial pressure, that is having the same reduction potential. The first issue is that the same O_2 partial pressure has been calculated based on an assumption that the water-gas-shift (WGS) reaction, reaction (12), is in thermodynamic equilibrium.

$$CO_2 + H_2 = CO + H_2O \tag{12}$$

It is reported that ^{6,10} the WGS reaction is in equilibrium above 500 °C as seen in Figure 14. As the total amount of oxygen is the same, that is the $\frac{CO+H_2}{CO_2+H_2O}$ ratio is constant in the gas, the reduction potential of the gas is quite constant even if the WGS reaction is not in equilibrium. The second, and more serious issue, is that C is being deposited according to the reverse reaction (5) ($2CO = C + CO_2$), especially when H₂ is present. This could be due to the WGS reaction which will give a higher CO/CO₂-ratio when H₂ is added. On a microscale it could also be due to the reactions (13) and (14), even though the equilibrium CH₄ partial pressure is $5 \cdot 10^{-5} bar$ at 800 °C for a 60/20/20 CO/H₂/CO₂ gas. The carbon deposition is determined by kinetics rather than the thermodynamics, and the CH₄ may decompose faster than the reaction rate of the reverse reaction 5 ($2CO = C + CO_2$).

$$2CO + 2H_2 = CH_4 + CO_2 \tag{13}$$



Figure 14. Shows the reaction quotients (Q) from the final gas composition of the hydrogen containing experiments and the equilibrium constant, Keq, from the water gas shift reaction, as a function of temperature. The equilibrium constants were calculated using HSC Chemistry¹⁰.

To summarize the impact of H_2 in CO/CO₂ mixtures, it is generally seen that H_2 will increase the reduction rate for the same reduction potential, that is the same or similar O₂ partial pressure. This is believed to be due to the mass transfer in the gas phase. As seen in Figure 15, the diffusion coefficient of H_2 in CO₂ is many times higher than the diffusion coefficient of CO in CO₂ and may hence explain the increased reaction rate. It has been shown in all the discussed studies that the prereduction of Mn-ores are not in equilibrium and hence the increased mass transfer with H_2 will increase the reaction rate.



Figure 15. The calculated binary diffusion coefficients for H₂-H₂O, H₂-CO₂ and CO-CO₂¹⁰

As a final short-point it can be mentioned that the decrepitation during reduction in H_2 containing gases has not changed compared to H_2 free gases¹⁰.

3 CONCLUSION

Today, solid carbon is used to partly pre-reduce the Mn-ores in the production of Mn-ferroalloys. The solid carbon can be substituted with H_2 and hence reduce the environmental footprint. In 4 studies it is shown that the H_2 containing gases will give higher reduction rate compared to CO/CO₂ gases with the same reduction potential. The reduction rate will be increased in the order of 20-100%, where H_2 has a larger impact on reduction rate for high oxygen pressures. In H_2 gas it is seen that the final degree of reduction is higher than using CO gas, and this

is believed to be due to the formation of metallic iron. The reduction rate in H_2 gases varies for various ores, and as for CO containing gases, the reduction rate of Comilog ore is faster than the Nchwaning ore, and activation energies of 23 kJ/mol versus 68 kJ/mol is respectively found for isothermal experiments. For the carbonate UMK ore, the presence of CO₂ in the gas, reduces the decomposition rate of the carbonate.

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