



Article Green and Efficient Utilization of Ferruginous Gibbsite Ore and Ferruginous Manganese Ore by Synergetic Carbothermic Co-Reduction–Magnetic Separation Process

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Abstract: The synergetic utilization of ferruginous gibbsite ores (Al-Fe ores) and ferruginous manganese ores (Mn-Fe ores) by the carbothermic co-reduction roasting–magnetic separation process was proposed as an innovative and green process for the separation and recovery of the valuable metal elements of Mn, Fe and Al from these ores. In this paper, a ferromanganese crude alloy with 72.47% Fe and 10.19% Mn and a high recovery of 85.89% Fe was prepared, which produces an acceptable feed to produce manganese steels with an electric arc furnace. The synergistic co-reduction of the two kinds of complex and refractory minerals was favored to separate Fe, Mn and Al from these ores. The influence of the operating variables on the recovery and separation of valuable metals from Mn-Fe ores and Al-Fe ores is initially studied. Then, the stepwise reduction behaviors of a composite oxide $Mn_{1-x}Fe_xO$ ($0 \le x \le 1$) and hercynite ($Mn_{1-y}Fe_yAl_2O_4$, $0 \le y \le 1$) were investigated to clarify that Mn-Fe ores have a positive impact on the reduction of fayalite and hercynite in Al-Fe ores. This study reported a simple green route, the carbothermic co-reduction–magnetic separation process, to economically and effectively treat Al-Fe ores and Mn-Fe ores.

Keywords: ferruginous gibbsite ores; ferruginous manganese ores; ferromanganese alloy; rich aluminum tailings; carbothermic co-reduction; magnetic separation

1. Introduction

Ferruginous gibbsite ores have received more attention as an alternative for high-grade alumina ores [1,2], and the national reserve of these ores in China attains to more than 1.5 billion tons [3,4], which are an important alumina resource that can meet the needs of the developing alumina industry. Unfortunately, of these, more than 80% of the Al-Fe ores with a higher iron content are acceptable as feedstocks in blast furnaces [5], which not only wastes the limited resources extravagantly, but also smelts the difficult high-alumina slag in the blast furnace [6].

In recent years, extensive research has been successively and comprehensively utilized on high alumina iron ores. Carbothermal roasting reduction is the essential procedure for recovering and separating aluminum and iron from Al-Fe ores. However, the formation of hercynite in the reduction process has limited the separation and recovery of the valuable metal elements of iron and aluminum from the Al-Fe ores [7,8]. Moreover, approximately 10% of the Al content and 20% of the Fe content was present in nonmagnetic and magnetic products, respectively [3,9,10], and thus the higher operating temperature and considerable energy consumption limited its industrial application. In addition, researchers have previously investigated the effect of nonmetallic or sodium salts on the separation of Al and Fe in Al-Fe ores, in which the recovery and grades of the Fe were all over 90% [7,9,11–14]; however, while the alkali salt was favored to the aggregation and growth of metallic iron particles, it was harmful to ironmaking and the utilization of iron, which limited the applica-



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Copyright: © 2022 by the authors. Licensee MDPI, Basel, Switzerland. This article is an open access article distributed under the terms and conditions of the Creative Commons Attribution (CC BY) license (https:// creativecommons.org/licenses/by/ 4.0/). tion of the process. It is urgent to put forward an efficient, innovative and environmentally friendly technology for recovering and separating aluminum and iron from Al-Fe ores.

Previously, we have proved that MnO has a positive impact on the reduction of fayalite and hercynite in Mn-Fe ores [15]. The thermodynamic theory calculations by the Factsage 7.0 software found that MnO can lower the reduction temperature of fayalite and hercynite [16–18], and the experimental results showed that the iron ions in the ferromanganese complex can be used as a reduction catalyst to decrease the reduction temperature of MnO to the ferromanganese alloy [18–20]. In addition, the Mn-Fe ores are an important manganese resource for manganese extraction, but they are unacceptable as feedstocks for smelting ferromanganese alloys [21–24]. The considerable research has found that the carbothermic reduction or magnetic roasting reduction via the magnetic separation process presented an effective and economical route to the separation and recovery of manganese and iron from Mn-Fe ores [25–30]. Although these technology is suitable for industrial applications, approximately 10~20% of the Mn element loss was separated into the magnetic products [20,24–26,30–33]. In summary, it can be inferred that Mn-Fe ores not only promoted the reduction of hercynite but also reinforced the reduction of the Al-Fe ores [34–36].

In this paper, an innovative, efficient and sustainable technology is proposed to synchronously utilize the Al-Fe ores and Mn-Fe ores by utilizing the carbothermic co-reduction roasting process–magnetic separation process. In this new cleaning technology, the main valuable metals of Fe, Mn and Al can be effectively and environmentally recovered from the Al-Fe and Mn-Fe ores, and the obtained magnetic product, namely, a ferromanganese alloy with a low Mn content and high Fe content, can be employed as feedstocks to produce manganese steels with an electric arc furnace. In addition, the obtained nonmagnetic products are rich aluminum slags that belong to the significant aluminum resources for aluminum extraction, which can satisfy the needs of the developing aluminum industry. Furthermore, this method could be extended to extract valuable metals from complex and refractory minerals.

2. Experimental Section

2.1. Raw Materials

The high iron gibbsite ore (Al-Fe ores) used in this research was obtained from Guangxi province in China, the low grade ferromanganese ore (Mn-Fe ores) was collected from South Africa and the reduction coal was obtained from Tianjin province in China. The Al-Fe ores comprise 34.68 wt.% TFe (total iron), 49.25 wt.% Fe₂O₃, 23.85 wt.% Al₂O₃ and 7.16 wt.% SiO₂, which exceeds the industrial high grade of natural ores. The XRD patterns of the raw materials (Figure 1a) displayed that the alumina elements in the Al-Fe ores mainly comprised gibbsite (Al(OH)₃), boehmite (AlOOH) and kaolinite (n(Al₂O₃)·SiO₂·n(H₂O)), and the iron compounds mainly existed in the form of goethite (FeOOH) and hematite (Fe₂O₃). As shown in Figure 1b, the Mn-Fe ore was a typical high-alumina ferromanganese ore, which comprised 29.50 wt.% TMn (total manganese), 28.28 wt.% TFe (total iron), and which mainly included manganite MnO(OH), bixbyite ((Fe,Mn)₂O₃ and hematite (Fe₂O₃). The bitumite coal contained 52.98 wt.% fixed carbon, 40.70 wt.% volatiles, 5.30 wt.% ash and 1.02 wt.% moisture.



Figure 1. XRD patterns of Fe-Al ores and Fe-Mn ores; (**a**) the XRD pattern of High alumina iron ore and (**b**) the XRD pattern of Low grade manganese iron ore.

2.2. Carbothermic Co-Reduction via Magnetic Separation Process

The schematic diagram of recovery and separation of valuable metal elements of Fe, Mn and Al from the Al-Fe ores and Mn-Fe ores is displayed in Figure 2. Figure 2a describes the schematic diagram of the separation and recovery of the valuable metal elements of Mn, Al and Fe from the Al-Fe ores and Mn-Fe ores by rotary kiln. Figure 2b shows that the simulated experiment was carried in a vertical resistance furnace. First, the Al-Fe ores and Mn-Fe ores as a raw material were dry-ground to particles sized below 74 µm. The given amount of the Mn-Fe ores, Al-Fe ores and the reduction coal were adequately mixed in a prescribed proportion of FC/O. It should be pointed out that the FC/O is the molar ratio of the fixed carbon in coal to the reducible oxygen of iron and manganese in the Mn-Fe ores and Al-Fe ores [6,9,15,20]. After that, under a pressure of 30 MP, the mixture with the addition of 10% moisture was briquetted into a cylinder sample with a diameter of 15 mm and height of 15 mm in a cylinder mold. Then, the briquettes were dried at 110 °C for 4 h in the vacuum drying oven. After that, the briquettes in a corundum crucible were put into the vertical resistance furnace at a fixed temperature for a present reduction roasting time. The reduced briquettes were then brought out and cooled to ambient temperature in Ar atmosphere. Our previous report also reported other detailed descriptions of the reduction roasting process [15,20,21,34].



Figure 2. (a) Schematic diagram of the industrial separation of Al, Mn and Fe from Fe-Al ores/Fe-Mn ores inside a rotary kiln; (b) Schematic diagram of experimental equipment for the carbothermic co-reduction process.

2.3. Magnetic Separation Process

The reduced briquettes were ground in a 2-MZ centrifugal grinding machine to pass through a 74 μ m sieve (100%), then the pulverous samples were magnetically separated by a DTCXG-ZN50 Davies magnetic tube to utilize the separation and recovery of the valuable metal elements of Mn, Al and Fe from the Mn-Fe ores and the Al-Fe ores. The magnetic product, named as a crude low grade ferromanganese alloy powder, can be directly used as an industrial raw material for the iron and steel industry, and magnetic products, also known as rich aluminum slag, can be directly used as industrial raw material for the aluminum industry. After the magnetic separation process, the separated samples were filtered and dried at 120 °C for 4 h in a vacuum drying oven. The valuable metal elements of Mn, Al and Fe of the separated product were measured by chemical titration at Northeastern University, Shenyang, China. Further detailed descriptions of the magnetic separation technological process were reported in our previous reports [15,20–22].

2.4. Separation Indexes

The following experimental assessment indicators were descripted as follows: the Fe recovery (R_{Fe}) and the Mn grade, the Fe grade and the Al₂O₃ content in the magnetic product. The Fe recovery of the magnetic separation can be calculated as follows:

$$R_{Fe} = \frac{m_{Fe} \times \beta_{Fe}}{M_{Fe} \times \gamma_{Fe}} \times 100\%$$
⁽¹⁾

where R_{Fe} is the recovery ratio of Fe, %; m_{Fe} is the mass of the magnetic product, g; M_{Fe} is the mass of the reduced samples, g; β_{Fe} is the Fe grade in the magnetic product, wt.%; γ_{Fe} is the Fe grade in the reduced samples, wt.%.

2.5. Analysis and Characterization Methods

After the reduction roasting process, the cooled samples were ground by a 200 mesh sieve by a grinder for X-ray diffraction (XRD, PANalytical B.V, X Pertpro, Almelo, Netherlands) detection. The XRD data were scanned by $5^{\circ} \sim 90^{\circ}$, step length 0.024° (20) and scanning speed 12° /min. Then, the phase composition and microstructure of the samples were determined by the XRD peak splitting software x' pert Highscore plus.

A scanning electron microscope (SEM, Jena, Germany) and energy dispersive spectrometer (EDS) were used to perform the point scanning on the microstructure of the products with different reduction roasting times. The interface reaction behavior in the synthetic process was also studied by SEM–EDS analysis.

Preliminary thermodynamic analysis of the sample was carried out by a synchronous thermal analyzer (DSC/DTA-TG, NETCH STA449 F3, Selb, Germany), and the preliminary results (DSC/DTA-TG) were comprehensively analyzed with the experimental results.

3. Thermodynamic Analysis of the Co-Reduced Al-Fe Ores and Mn-Fe Ores

Figure 3 displays the thermodynamic theory calculations and thermogravimetric analysis of the Al-Fe ores and the Mn-Fe ores to clarify the co-reduction mechanism of the Al-Fe ores and Mn-Fe ores. Figure 3a displays the thermodynamic theory calculation analysis of iron oxide and manganese oxide in the Al-Fe ores and Mn-Fe ores using the FACTSAGE 7.2 software. The main chemical reaction equations obtained are Equations (2)–(11). The carbothermic co-reduction mechanism of the Mn-Fe ores and Al-Fe ores has been studied in our previous research [15,20,21]. The poor separation of Fe and Al from the Al-Fe ores is mainly due to the unreduced hercynite phase and unreduced fayalite phase. The reduction temperature of the hercynite (FeAl₂O₄) and fayalite (Fe₂SiO₄) was 841.5 $^{\circ}$ C and 815.6 °C under the CO atmosphere, respectively. In the presence of MnO, the indirect reduction temperature of FeAl₂O₄ and Fe₂SiO₄ decreased to 658.6 °C and 667.9 °C under CO atmosphere, respectively. In addition, the partial pressure of the CO required for the reduction reaction was also significantly reduced. As shown in Figure 3b, in the presence of MnO, the direct reduction temperature of FeAl₂O₄ obviously decreased from 831 °C to 667 °C, and that of Fe₂SiO₄ decreased from 805 °C to 658 °C. The main chemical reaction equations obtained are Equations (12)–(15). It was proved that the participation of MnO in the Fe-Mn ores can significantly lower the reduction temperature of hercynite ($FeAl_2O_4$) and fayalite (Fe_2SiO_4) in the Al-Fe ores.

The thermogravimetric analysis of the Al-Fe ores and Mn-Fe ores is described in Figure 3c to clarify the different reduction stages of the Mn-Fe ores and Al-Fe ores. In the carbothermic reduction process of the Mn-Fe ores, the reduction stage A1 (Mn_xO_y \rightarrow MnO) had priority over the reduction stage A2 (Fe_xO_y \rightarrow FeO). The reduction stage A3 of FeAl₂O₄ and Fe₂SiO₄ in the presence of MnO occurred before the reduction stage A4 (FeAl₂O₄ and Fe₂SiO₄ \rightarrow Fe). In addition, in the carbothermic reduction process of the Al-Fe ores, the crystal water precipitation in the Al-Fe ores also initially occurred before the other reduction stage B3 (FeAl₂O₄ \rightarrow Fe). In the process of the carbothermic co-reduction of the mixed samples containing 50% Mn-Fe ores and 50% Al-Fe ores, by focusing on the co-reduction mechanism of the two ores, the reduction stage C2 of FeAl₂O₄ in the presence of MnO had priority over the reduction stage C3 (FeAl₂O₄ \rightarrow Fe).

$$Fe_3O_4 + CO = Fe + CO_2 \tag{2}$$

$$FeO + CO = Fe + CO_2$$
(3)

 $Fe_3O_4 + CO = FeO + CO_2 \tag{4}$

$$Al_2O_3 + Fe_2O_3 = Fe_2Al_2O_6$$
(5)

$$C + CO_2 = 2CO \tag{6}$$

 $FeAl_2O_4 + MnO + CO = Fe + MnAl_2O_4 + CO_2$ (8)

$$Fe_2SiO_4 + 2CO = SiO_2 + 2Fe^+ + 2CO_2$$
 (9)

$$Fe_2SiO_4 + 2MnO + 2CO = 2Fe + Mn_2SiO_4 + 2CO_2$$
 (10)

$$Mn_3O_4 + CO = 3MnO + CO_2 \tag{11}$$

$$Fe_2SiO_4 + 2C = SiO_2 + 2Fe^+ + 2CO$$
 (12)

$$Fe_2SiO_4 + 2MnO + 2C = 2Fe + Mn_2SiO_4 + 2CO$$
(13)

$$FeAl_2O_4 + C = Al_2O_3 + Fe + CO$$
(14)

$$FeAl_2O_4 + MnO + C = Fe + MnAl_2O_4 + CO$$
(15)



Figure 3. (a) Gas–equilibrium diagram of Mn-Fe-C-O system; (b) ΔG^0 -T of FeAl₂O₄ and Fe₂SiO₄; (c) TG/DTG of the Fe-Mn ores and Fe-Al ores from ambient temperature to 1100 °C at a heating rate of 10 °C/min in a pure Ar atmosphere.

As shown in Figure 4, with the reduction time increasing from 5 min to 30 min, the metallization rate increased from 39.04% to 87.34% at 1300 °C and from 21.11% to 49.05% at 1150 °C in the carbothermic reduction process of Al-Fe ores, but when the Al-Fe ores were mixed with the Mn-Fe ores for the carbothermic reduction roasting process, the metallization rate increased from 61.90% to 99.78% at 1300 °C and from 36.18% to 80.40% at 1150 °C in the carbothermic co-reduction roasting process of the Al-Fe ores and Mn-Fe ores. In summary, the TG/DTG analysis results and the metallization rate for the reduced Mn-Fe ores and Mn-Fe ores were consistent with the thermodynamic theory calculations. In other words, it proves once again that the participation of MnO can obviously lower the reduction temperature of hercynite (FeAl₂O₄) and fayalite (Fe₂SiO₄) in Al-Fe ores. The presence of MnO can promote the reduction of Al-Fe ores, but the mechanism analysis of bauxite is not clear in the presence of MnO, so it is necessary to further study its reduction mechanism.



Figure 4. Effect of reduction time and temperature on metallization rate at a FC/O of 1.1 in reduction roasting process of (**a**) Fe-Al ores and (**b**) 70%Fe-Mn ore + 30% Fe-Mn ore.

4. Results and Discussion

4.1. Results of the Carbothermic Co-Reduction Roasting–Magnetic Separation of Al-Fe Ores/Mn-Fe Ores

To separate the valuable metal elements of Fe and Al from the Al-Fe ores, the carbothermic reduction roasting of the reduced product was employed to transform iron oxides and manganese oxides into the ferromanganese alloy. The magnetic product, as the ferromanganese alloy, and the nonmagnetic product, as the rich aluminum slag, were recovered by the magnetic separation process. The effect of the reduction temperature, FC/O, reduction time and proportion of the Al-Fe ores/Mn-Fe ores on the recovery and grade of the Fe and Mn grade in smelting the ferromanganese alloy are displayed in Figure 5. As shown in Figure 5a, with the reduction temperature increasing from 1200 °C to 1400 °C, the Fe grade in the ferromanganese alloy continuously increased from 48.73% to 76.70%, and the Al₂O₃ content decreased from 21.49% to 3.23%, while the Mn grade in the alloy remained at around 10.0%. These proved that the higher reduction temperature was favorable to the reduction of hercynite in the Al-Fe ores. In Figure 5b, it can be found that the Fe recovery rapidly decreased from 92.50% to 69.21%, then slowly increased to 74.68% with the increase of FC/O from 0.7 to 1.2. The Mn grade in the alloy increased from 6.20% to 11.18%. That is to say, FC/O is not conductive to the recovery of Fe, but it is benefited in the reduction of manganese oxides. It is well known that both the reduction of iron oxides and manganese oxides is enhanced with the rise in FC/O [6,15,20]. Thus, it was deduced that there was a certain relationship between the reduction of iron and the manganese oxides. In Figure 5c, with the increase of the reduction time from 1 h to 2 h, the recovery of Fe increased from 79.97% to 89.35%, the Fe and Mn grade in alloy increased from 67.72% to 72.47% and from 8.93% to 10.19%, respectively. When the reduction time exceeded 2 h, the Fe and Mn grade in the alloy remained at 10.2% and 72.0%, respectively. As shown in Figure 5d, it can be seen that with the increase of the proportion of the Al-Fe ores/Mn-Fe ores from 5/5 to 9/1, the Fe recovery increased from 63.93% to 92.14%, and the grade of Fe in the alloy increased from 66.55% to 73.54%. The Mn grade in the alloy decreased due to the decrease of the Mn-Fe ores ratio. Under the optimum conditions, the raw Mn-Fe ores and Al-Fe ores were reduced at 1300 °C for 2 h under the FC/O of 1.0 with the proportion of Al-Fe ores/Mn-Fe ores of 7/3, which obtained a crude Mn-Fe alloy containing the Fe grade of 72.47%, the Mn grade of 10.19% and the Fe recovery of 89.35%, which can be used in an electronic furnace to produce manganese steels with an electric arc furnace.



Figure 5. Effect of carbothermic co-reduction temperature (**a**), FC/O (**b**), reduction time (**c**) and proportion of Fe-Al ores/Fe-Mn ores (**d**) on separation indexes of Al, Mn and Fe from Fe-Al ores/Fe-Mn ores.

4.2. Phase Transformation of Al-Fe Ores and Mn-Fe Ores during Carbothermic Co-Reduction Roasting

4.2.1. Phase Transformation of Al-Fe Ores in Reduction Roasting Process

Figure 6 displays the XRD analysis of the Al-Fe ores reduced at 1300 °C under the FC/O of 1.0 for a duration time of 5–90 min. When the reduction time reached 5 min, the characteristic diffraction peaks of Fe, FeAl₂O₄, Fe₂SiO₄ and SiO₂ can be found. With the reduction time increased from 5 min to 90 min, the intensity of the characteristic diffraction peaks of Fe₂SiO₄ gradually decreased, and that of SiO₂ and Al₂SiO₅ increased, which demonstrated that almost all of the Fe₂SiO₄ was converted to SiO₂ and Al₂SiO₅. In addition, it was also observed that the intensity of the diffraction peaks of hercynite FeAl₂O₄ gradually decreased, and that of Fe increased, which demonstrated that FeAl₂O₄ was reduced to Fe and Al₂O₃. When the reduction reached 60 min, Al₂O₃ and SiO₂ combined to form Al₂SiO₅. When the time reached 90 min, the amount of FeAl₂O₄ was already very small. However, the presence of FeAl₂O₄ was accompanied by the entire experimental process. Therefore, the reduction of hercynite FeAl₂O₄ was restricted to the reduction form Al-Fe ores during the reduction roasting process.



Figure 6. XRD patterns of the Fe-Al ores reduced at 1300 °C under FC/O of 1.0 for different reduction roasting times.

4.2.2. Phase Transformation of Mn-Fe Ores in Reduction Roasting Process

Figure 7 shows the XRD patterns of the Mn-Fe ore reduced for 30 min under the FC/O of 1.0 at different reduction temperatures. As shown in Figure 7a, the intensity of the characteristic diffraction peaks of the Fe gradually increased with the rise of the reduction temperature from 800 °C to 1000 °C, and the characteristic diffraction peaks of Fe-C were detected at 1100 °C. The standard substance of the characteristic diffraction peaks of the Fe-C group are found in Figure 7d. Figure 7b shows the evolution rule of the characteristic diffraction peaks of wustite (Fe_xMn_{1-x}O, $1 \ge x \ge 0$) in the carbothermal reduction process of the Mn-Fe ores. The standard substance of the characteristic diffraction peaks of the wustite $Fe_xMn_{1-x}O$ group are also displayed in Figure 7g. It was also observed that the characteristic diffraction peaks (2 0 0) of the wustite $Fe_xMn_{1-x}O$ transformed from 41.802 (FeO) to 40.663 (MnO), which was detected as the x-values increased from 1 to 0 with the reduction temperature rising from 800 °C to 1100 °C. Moreover, the intensity of the diffraction peaks of the wustite Fe_xMn_{1-x}O group remained at a level value with the increase of the reduction temperature. It was inferred that the wustite $Fe_xMn_{1-x}O$ phase was continually generated from the formation of a composite oxide of Fe oxide and Mn oxide. Figure 7c displays the XRD patterns of the fayalite $Fe_vMn_{2-v}SiO_4$ and hercynite $Fe_vMn_{1-v}Al_2O_4$ group in the carbothermic reduction process of the Mn-Fe ores. The standard substance of characteristic diffraction peaks of the fayalite $Fe_vMn_{2-v}SiO_4$ and the hercynite $Fe_vMn_{1-v}Al_2O_4$ are illustrated in Figure 7e,f. As shown in Figure 7c, with the rising of the reduction temperature, the intensity of the characteristic diffraction peaks of the fayalite Fe_vMn_{2-v}SiO₄ and the hercynite Fe_vMn_{1-v}Al₂O₄ gradually increased, and the transitional phenomenon for the characteristic diffraction peaks of the fayalite Fe_vMn_{2-v}SiO₄ was also observed. However, it was difficult to distinguish the y-value of the $Fe_{v}Mn_{2-v}SiO_{4}$ and $Fe_{v}Mn_{1-v}Al_{2}O_{4}$ groups according to the standard substance of these groups. According to the previous study, the ionic Fe^{2+} in hercynite (FeAl₂O₄) and fayalite (Fe₂SiO₄) was easily replaced by ionic Mn^{2+} in MnO to form Fe_vMn_{2-v}SiO₄ and $Fe_vMn_{1-v}Al_2O_4$, which promoted the stepwise co-reduction of fayalite $Fe_vMn_{2-v}SiO_4$ and $Fe_vMn_{1-v}Al_2O_4$ in the Mn-Fe ores and Al-Fe ores [35,36]. The formation of the wustite $Fe_xMn_{1-x}O$ phase was closely related to the stepwise co-reduction of fayalite $Fe_yMn_{2-y}SiO_4$ and $Fe_vMn_{1-v}Al_2O_4$ [37,38]. The close combination of the MnO phase and a metallic Mn-Fe alloy that derived from the stepwise reduction of the wustite $Fe_xMn_{1-x}O$ phase were



closely related with the stepwise reduction of the hematite (Fe, $Mn)_2O_3$ phases and fayalite $Fe_yMn_{2-y}SiO_4$, hercynite (Fe_yMn_{1-y}) Al_2O_4 .

Figure 7. XRD patterns of Fe-Mn ores reduced for 30 min under FC/O of 1.0 at different reduction temperature, (**a**) is the XRD pattern of $2\theta = 43 \sim 2\theta = 49$; (**b**) is the XRD pattern of $2\theta = 34.1 \sim 2\theta = 43.3$; (**c**) is the XRD pattern of $2\theta = 30 \sim 2\theta = 38$ and XRD patterns of standard substances of the (**d**) Fe_xC group, (**e**) Fe_yMn_{2-y}SiO₄ group, (**f**) (Fe_yMn_{1-y})Al₂O₄ group and (**g**) (FeO)_x(MnO)_{1-x} group.

4.2.3. Phase Transformation of Al-Fe Ores/Mn-Fe Ores

Figure 8 shows the effect of the different reduction roasting times on the phase transformations of the reduced mixed samples at 1300 °C under the FC/O of 1.0 with the proportion of the Al-Fe ores/Mn-Fe ores of 7/3. It was initially found that Fe, FeAl₂O₄, Fe_{0.331}Mn_{0.669}O and FeMn(SiO₄) were the major phase compositions in the Mn-Fe ores. With the reduction time increased from 5 to 90 min, the diffraction intensity peaks of the metallic Fe initially increased, then slowly decreased, which demonstrated that the formation of metallic iron was converted to the Fe-C alloy. A complex transformation phenomenon for the characteristic diffraction peaks (2 0 0) of the wustite Fe_xMn_{1-x}O from 41.802 (FeO) to 40.663 (MnO) was detected as the x-values decreased from 1 to 0 while

the reduction time increased from 5 to 90 min. Stabilization was broken until the reaction proceeded to 60 min. Due to the phenomenon of carburization, the Fe content was lowered to produce $C_{0.14}Fe_{1.86}$. It can be seen from the XRD analysis that the peak of FeAl₂O₄ is constantly shifted toward the peak of MnAl₂O₄. This shows that FeO in FeAl₂O₄ is gradually replaced by MnO. It maintains the lattice structure of hercynite and exists in the form of Mn_xFe_{1-x}Al₂O₄. As the reduction time increased, the value of x gradually increased. When the experiment was carried out for 90 min, its peak was very close to MnAl₂O₄. This means that a part of MnAl₂O₄ has been generated at this time, and the amount of FeAl₂O₄ was already very small.



Figure 8. XRD patterns of the reduced mixed samples at 1300 °C under FC/O of 1.0 with proportion of Fe-Al ores/Fe-Mn ores of 7/3 for different reduction roasting times.

The reduced samples were detected by XRD analysis to reveal the co-reduction mechanism of the Al-Fe ores and Mn-Fe ores. Figure 9 shows the effect of the proportion of the Al-Fe ores/Mn-Fe ores on the phase transformation of the reduced mixed samples at 1300 °C for 2 h under FC/O of 1.0. As shown in Figure 9a, where it can be found that fayalite was the major mineral in the reduced samples, with the increase of the proportion of the Al-Fe ores/Mn-Fe ores from 9/1 to 5/5, the intensity of the diffraction peaks of Fe and Al_2O_3 decreased and disappeared, and that of metallic iron also gradually decreased; these also demonstrate that it is much more difficult to reduce the hercynite FeAl₂O₄ in the Al-Fe ores, and in the presence of MnO, the Mn-Fe ores were favored to the stepwise reduction of $FeAl_2O_4$ in the Al-Fe ores. In Figure 9b, the intensity of the diffraction peaks of the hercynite $(Fe_vMn_{1-v})Al_2O_4$ gradually increased. When the proportion of the Al-Fe ores/Mn-Fe ores reached to 5/5, the intensity of the diffraction peaks of the hercynite (Fe_vMn_{1-v}) Al_2O_4 reached its maximum. Moreover, the diffraction peaks of Al₂O₃ and the wustite Fe_xMn_{1-x}O disappeared, and the diffraction peaks of the Mn-Fe-C alloy appeared. The above results further confirm that MnO in Mn-Fe ores can catalyze the stepwise reduction of hercynite FeAl₂O₄ in Al-Fe ores to form metallic iron.



Figure 9. XRD patterns of the reduced briquettes at 1300 °C for 2 h under FC/O of 1.0 with the different proportions of Fe-Al ores/Fe-Mn ores, (**a**) is the XRD pattern of $2\theta = 41.7 \sim 2\theta = 47.5$ and (**b**) is the XRD pattern of $2\theta = 29.8 \sim 2\theta = 38.1$.

4.3. Morphological Characteristics of Al-Fe Ores and Mn-Fe Ores during Carbothermic Reduction Process

To further confirm these conclusions, Figure 10 displays that the SEM/EDS were used in the microstructure of the reduced samples. The white phase was mainly metallic iron. A mixture of FeAl₂O₄, SiO₂, and so on, constitutes the gray base phase in high alumina iron ore. It can be clearly seen from the EDS detecting results that the gray phases in the mixed sample are mainly MnAl₂O₄ and FeAl₂O₄ and their transition phases. A trend of increasing particle sizes of the ferromanganese alloy with the increasing reduction time was evident. Compared with the Al-Fe ores and the Al-Fe ores/Mn-Fe ores, the additions of the Mn-Fe ores resulted in a significantly higher average and maximum of ferromanganese particle sizes. As shown in Figure 11, in the case of the pure Al-Fe ores, the metal iron formed by the reduction of the high alumina iron ore is approximately spherical. Iron particles grow slowly and are difficult to aggregate. Compared with the Al-Fe ores and the Al-Fe ores/Mn-Fe ores, the degree of aggregation and the amount of metallic iron in the mixed samples reduction product are significantly increased. In the reduction process, there is a transition state in which Fe, Mn, and Al coexist. After the transition state, the iron particles grow rapidly and aggregate into an irregular shape. These proved that the MnO element acts as reduction catalyst to aid in the growth of the ferromanganese alloy particles.



Figure 10. Micrographs of the reduced samples at 1200 °C under FC/O of 1.0 for different reduction times.



Figure 11. SEM phase analysis for the reduced samples at 1200 °C under FC/O of 1.0 for different reduction times. (**a**) and (**b**) respectively represent The point sweep spectra of High Alumina Iron Ore and The mixed sample with a reduction time of 30 min, (**c**) and (**d**) respectively represent High Alumina Iron Ore and The mixed Sample Point scan spectrum with the reduction time of 90 min.

4.4. Carbothermic Co-Reduction Mechanism of Al-Fe Ores and Mn-Fe Ores

Figure 12 shows the carbothermic co-reduction mechanism of the Al-Fe ores and the Mn-Fe ores. As illustrated in Figure 12a, based on the experimental results and the thermodynamic theory calculations, the main phase consists of a spinel-type phase, a monoxide and an iron phase. Moreover, the Al-spinel refers to $FeAl_2O_4$, $(Fe_yMn_{1-y})Al_2O_4$ and $MnAl_2O_4$. The formation of the spinel phase types are favored by the theoretical calculation

of thermodynamics. Figure 12b shows the crystal structure of the spinel (Fe_yMn_{1-y}) Al_2O_4 . The spinel FeAl₂O₄ had normal and inverse spinel structures, and the distribution of Al^{3+} , Fe^{2+} and Mn^{2+} ions in the spinel structure were disordered at higher temperatures. The normal and inverse spinel structure of the spinel FeAl₂O₄ samples inevitably coexisted and were co-transformed at the same time. Therefore, it provides a comfortable physical and chemical environment for the reduction of hercynite FeAl₂O₄ in the Al-Fe ores. In the presence of MnO, the Mn^{2+} can replace Fe^{2+} easily in FeAl₂O₄, resulting in the formation of MnAl₂O₄ and FeO, which further promotes the reduction of Fe. In addition, the free Fe^{2+} ions were combined with MnO to form the wustite ($Fe_xMn_{1-x}O$), which further reduced to ferromanganese alloy.



Figure 12. (**a**) Ternary phase diagram of Fe₂O₃-MnO₂-Al₂O₃ system; (**b**) Crystal structure of the spinel FeAl₂O₄ and MnAl₂O₄; (**c**) Phase transformation of FeAl₂O₄ in carbothermic co-reduction process.

Figure 12c displays the phase transformation of the hercynite FeAl₂O₄ in the reduction process of the Al-Fe ores and the Mn-Fe ores. The situation A was the formation of MnAl₂O₄ according to Equations (16)–(18), and in the presence of MnO, the FeAl₂O₄ was converted to (Fe_yMn_{1-y})Al₂O₄, and further converted to MnAl₂O₄. In addition, the replaced iron atoms were easily reduced to metallic iron at 685.6 °C in a CO atmosphere. The XRD analysis (Figures 8 and 9) proved that the Fe²⁺ atoms were indirectly replaced by Mn²⁺ atoms in the spinel-type structure of FeAl₂O₄, resulting in the formation of Fe_yMn_{1-y}Al₂O₄. The situation B shows that the reduction temperature of hercynite FeAl₂O₄ was 841 °C in a CO atmosphere or at 831 °C by carbon (Equations (19) and (20)), respectively. The situation C was the formation of the wustite (Fe_xMn_{1-x}O) according to Equations (21)–(24), and in

the presence of MnO, the Fe²⁺ atoms that were substituted by Mn^{2+} atoms in the crystal structure of spinel (Fe_yMn_{1-y})Al₂O₄ were easily able to combine with the free MnO to form the new phase of Fe_xMn_{1-x} O. Then, it was further reduced to metallic iron by carbon at 724 °C and by CO at 731 °C. These demonstrated that MnO within Mn-Fe ores acted as a catalyst to promote the reduction of hercynite FeAl₂O₄ in the Al-Fe ores, and thus improve the metallization rate of Fe and Mn. The stepwise reduction reactions of FeAl₂O₄ by MnO were expressed as follows:

$$[Fe^{2+}][Al^{3+}]_2[O^{2-}]_4 + (1-y)[Mn^{2+}][O^{2-}] = [Fe^{2+}]_y[Mn^{2+}]_{1-y}[Al^{3+}]_2[O^{2-}]_4 + (1-y)[Fe^{2+}][O^{2-}]$$
(16)

$$[Fe^{2+}]_{y}[Mn^{2+}]_{1-y}[Al^{3+}]_{2}[O^{2-}]_{4} + y[Mn^{2+}][O^{2-}] = [Mn^{2+}][Al^{3+}]_{2}[O^{2-}]_{4} + y[Fe^{2+}][O^{2-}]$$
(17)

$$[Fe^{2+}][Al^{3+}]_2[O^{2-}]_4 + (2-y-x)[Mn^{2+}][O^{2-}] \rightarrow$$

$$[Fe^{2+}]_y[Mn^{2+}]_{1-y}[Al^{3+}]_2[O^{2-}]_4 + [Fe^{2+}][O^{2-}]_x[Mn^{2+}][O^{2-}]_{1-x}$$
(18)

$$[Fe^{2+}][Al^{3+}]_2[O^{2-}]_4 + CO = [Fe] + [Al^{3+}]_2[O^{2-}]_3 + CO_2$$
(19)

$$[Fe^{2+}][Al^{3+}]_2[O^{2-}]_4 + C = [Fe] + [Al^{3+}]_2[O^{2-}]_3 + CO$$
(20)

$$y[Fe^{2+}][O^{2-}] + (1-y) [Mn^{2+}][O^{2-}] = [Fe^{2+}][O^{2-}]_y[Mn^{2+}][O^{2-}]_{1-y}$$
(21)

$$[Fe^{2+}][O^{2-}]_{y}[Mn^{2+}][O^{2-}]_{1-y} + [C^{2+}][O^{2-}] = y[Fe] + 1 - y[Mn^{2+}][O^{2-}] + [C^{2+}][O^{2-}]_{2}$$
(22)

$$[Mn^{2+}][O^{2-}] + [Fe][C] = [Fe] + [Mn] + [C^{2+}][O^{2-}]$$
(23)

$$[Fe^{2+}][O^{2-}] + [C^{2+}][O^{2-}] = [Fe] + [C^{2+}][O^{2-}]_2$$
(24)

5. Conclusions

The synergetic utilization of ferruginous gibbsite ore and ferruginous manganese ore by carbothermic co-reduction followed by the magnetic separation process is reported as an effective technological process to separate and recover the valuable metal elements of Mn, Fe and Al from these ores, which produces an acceptable feed to produce manganese steels in an electronic furnace and the aluminum electrolysis industry.

A ferromanganese crude alloy with 72.47% Fe and 10.19% Mn was prepared under the appropriate conditions of a co-reduction temperature (1300 °C), a co-reduction duration (2 h), an FC/O (1.0), a proportion of Al-Fe ores/Mn-Fe ores (7/3) and a magnetic intensity (75 mT), which produced an acceptable feed to produce manganese steels with an electric arc furnace. The corresponding recovery of Fe can reach 89.35%. Meanwhile, the Al element from the Al-Fe ores and the Mn-Fe ores was enrichened into the nonmagnetic tailing, which can be extracted easily by the aluminum electrolysis industry.

The Mn-Fe ores were found not only to promote the reduction of hercynite but also to reinforce the reduction of the Al-Fe ores, which improved the metallization rates and the rapid formation of the ferromanganese alloy in the reduced samples. The FeAl₂O₄ in the Al-Fe ores was reduced to metallic iron at 841 °C in a CO atmosphere or at 831 °C by carbon. In the presence of MnO, FeAl₂O₄ in the Al-Fe ores was reduced to metallic iron at 731 °C in a CO atmosphere or at 724 °C by carbon. The MnO within the Mn-Fe ores acted as a catalyst to promote the reduction of hercynite FeAl₂O₄, and thus improved the metallization rate of Fe in the Al-Fe ores.

The synergistic co-reduction of the two kinds of complex and refractory minerals was favored to separate Fe, Mn and Al from these ores in the carbothermic reduction process. The formation mechanism and stepwise reduction of hercynite $FeAl_2O_4$ was processed as follows: $FeAl_2O_4$ was directly reduced to metallic iron in CO and C; $FeAl_2O_4$ was converted to $(Fe_yMn_{1-y})Al_2O_4$ and further converted to $MnAl_2O_4$; the Fe^{2+} atoms that were replaced by Mn^{2+} atoms in the stepwise reduction of hercynite $FeAl_2O_4$ were easily combined with free MnO to form a new ferromanganese composite phase ($FeO_xMn_{1-x}O$), which was further reduced to the ferromanganese alloy.

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