

Design of Nonlinear Droop Control in DC Microgrid for Desired Voltage Regulation and Current Sharing Accuracy

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Abstract—It is well known that there is a design tradeoff between voltage regulation and load current sharing in conventional linear droop control for several parallel-connected distributed sources in a DC microgrid. To improve both performance areas, this article proposes a design method to identify the nonlinear droop curve coefficient with the desired bus voltage regulation and current sharing accuracy in a specified heavy load range, while a linear droop function with a negative droop resistance is used to tighten the bus voltage regulation in the light load range. The design procedure of load boundary and the effect of negative droop resistances on the system stability are analyzed in detail. Also, a boost-up and curve-fitting method can further improve the voltage regulation and achieve smooth transition between light load and heavy load. In this way, no additional voltage compensation loop is needed. Two parallel-connected DC converters are finally built to verify the theoretical analysis.

Index Terms—DC microgrid, voltage regulation, load current sharing, nonlinear droop control, curve fitting.

I. INTRODUCTION

CURRENTLY, renewable sources account for an increasing proportion of the total electricity generation in the world. Meanwhile, more and more DC equipments are connected to the grid, such as DC-powered data centers, LED lighting systems, electric vehicle charging stations, and so on. The DC grid is simpler and more efficient considering DC ports of both the source and the load [1], [2]. Therefore, the DC grid becomes popular in recent years. Fig. 1 shows a typical configuration of DC microgrid including multiple solar arrays, loads, and the energy storage station [3], [4]. With different DC voltage levels of sources and loads, interfacing converters need to connect a single DC bus for sources and loads.

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When multiple sources are paralleled into a DC bus, the bus voltage regulation and load sharing accuracy become crucial issues due to different output impedances of converters and line resistances among different converters and the DC bus [5]–[7]. To achieve a tight voltage regulation and an accurate load current sharing, there are some existing methods, which can be classified into centralized and decentralized methods [8], [9]. It is well-known that centralized methods adopt one central controller to adjust the bus voltage and send the same current sharing signals to other converters for power balance [10]–[12]. However, if the central controller fails to run, the DC microgrid will collapse. Besides, the precise communication link is of most importance in the centralized method. To save the communication link and improve system reliability, decentralized methods are expected to find more and more applications, where the droop control is a popular method to construct relatively large virtual resistances in the circuit, thus reducing the load imbalance among different sources [13]–[17]. Obviously, the large droop resistances will make the bus voltage drop even below the minimum bus voltage limit. Some works use hierarchical control by increasing the voltage reference to compensate the bus voltage drop [18]–[20]. But the hierarchical control still needs communication feedback and increases the system complexity. So it is challenging for the linear droop control to achieve good performances for both bus voltage regulation and load sharing.

Considering a practical light loading condition, the output current of each source converter is well below the current limit and the accuracy of load sharing is not an issue for this case. Then the droop resistance can be designed small to permit a tight bus voltage regulation. At heavy load conditions, the interfacing converters suffer almost the full ratings and a small current imbalance may cause over-loading or even failure of some converters. Thus, the droop resistance at heavy loads is expected to be large enough to improve load sharing accuracy. Based on these considerations, some nonlinear droop control methods have been proposed to satisfy the requirements of both load sharing and bus voltage regulation. Reference [21] uses an inverse parabolic function to improve the bus voltage regulation at light loads and to permit good current sharing at heavy loads. But it is hard to design the converters with the desired voltage regulation and current sharing accuracy. Reference [22] gives a nonlinear curve to

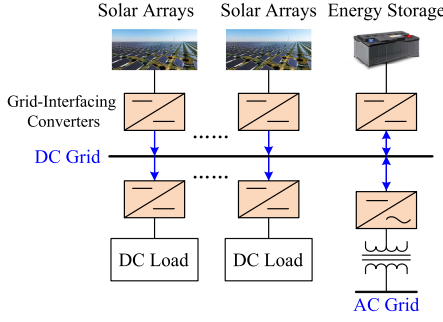


Fig. 1. A typical configuration of a DC microgrid.

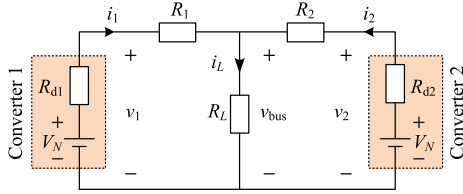


Fig. 2. A simplified DC microgrid with two parallel-connected converters. The shadow area shows equivalent circuits of two converters.

satisfy both requirements, where the droop curve coefficients at different loads are hard and complicated to design. Reference [23] combines droop curves with odd coefficient n , i.e., $n = 1, 3, 5, \dots$, into a polynomial equation as the droop function. However, the bus voltage still exceeds the limit at some loads and a voltage compensation loop is needed.

To solve the above issues, this article proposes a simple design method to identify the nonlinear droop curve coefficient with the desired bus voltage regulation and current sharing accuracy in a specified heavy load range. In the light load range, a linear droop function with a negative droop resistance is proposed to tighten the bus voltage regulation with the allowable current sharing accuracy. In this way, no additional voltage compensation loop is needed. Besides, a boost-up for heavy load and a curve-fitting for the whole load can further improve the bus voltage regulation and ensure the smooth transition between light load and heavy load. The design method can be readily extended to the DC microgrid system with multiple parallel-connected converters. The system stability with the proposed nonlinear droop control is also analyzed considering variations of output load and cable impedances. Finally, two parallel-connected DC converters are built to verify the theoretical analysis.

II. FUNDAMENTAL ANALYSIS OF PROPOSED NONLINEAR DROOP CONTROL DESIGN

To simplify the analysis, a simple DC microgrid system consisting of two parallel-connected converters is applied to feed a load directly, as shown in Fig. 2. Here, $v_{1,2}$ and $i_{1,2}$ are the output voltages and currents of Converter 1 and Converter 2, while R_1 and R_2 are the line resistances between the converter output port and the load R_L . v_{bus} is the DC bus voltage.

To provide the accurate load sharing performance, the conventional droop control constructs constant droop resistances R_{d1} and R_{d2} in series with R_1 and R_2 , and then parallel-connected converters can be further modeled as a

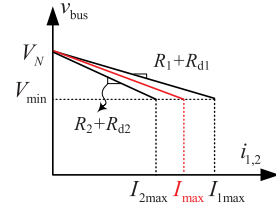


Fig. 3. Bus voltage varying with converter currents under linear droop control.

no-load voltage connected with the droop resistance, as shown in the shadow area of Fig. 2. Usually, the parallel-connected converters are designed at the same no-load voltage V_N . Given the acceptable bus voltage minimum V_{min} and converter output current maximum I_{imax} ($i = 1$ or 2), the linear droop resistances are designed as

$$R_{di} = \frac{V_N - V_{min}}{I_{imax}}. \quad (1)$$

Then the actual output voltage reference v_{refi} of converters becomes

$$v_{refi} = V_N - R_{di}i_i. \quad (2)$$

When (2) is implemented as the reference of $v_{1,2}$ in the voltage feedback PI controller, the linear droop control makes converter current distributed in (3).

$$\frac{i_1}{i_2} = \frac{R_2 + R_{d2}}{R_1 + R_{d1}}. \quad (3)$$

Define the current sharing error e_I as

$$e_I = \left| \frac{i_1 - i_2}{i_1 + i_2} \right| = \left| \frac{R_1 - R_2 + R_{d1} - R_{d2}}{R_1 + R_2 + R_{d1} + R_{d2}} \right|. \quad (4)$$

From (4), to improve the load sharing accuracy, i.e., to reduce e_I , R_{d1} and R_{d2} are expected to be close, which means two converters have the identical current rating I_{imax} as shown in Fig. 3. Besides, the larger R_{d1} and R_{d2} are, the smaller e_I is. But the linear droop function in (1) has no capability to enhance $R_{d1,2}$ with the given voltage drop. Therefore, to further reduce e_I , the nonlinear droop function with a high curve coefficient is recommended.

A. Identification of Nonlinear Droop Curve Coefficient for Heavy Load

The general nonlinear droop curve can be expressed as follows [22], where n is the droop curve coefficient and $n > 1$.

$$v_{refi} = V_N - k_i i_i^n, \text{ where } k_i = \frac{V_N - V_{min}}{I_{imax}^n}. \quad (5)$$

Then the equivalent droop resistance R_{di} working at a load current of I_i can be derived as

$$R_{di}|_{i_i=I_i} = \left| \frac{dv_{refi}}{di_i} \right|_{i_i=I_i} = nk_i i_i^{n-1}|_{i_i=I_i}. \quad (6)$$

From (6), equivalent droop resistances vary with the operating load and droop coefficient, as demonstrated in Fig. 4 with the example case of $V_N = 48$ V, $V_{min} = 46.8$ V, i.e., the voltage regulation ratio $e_V = \frac{V_N - V_{min}}{V_N} = 2.5\%$, and $I_{1max} = I_{2max} = 6$ A. It can be seen that R_{di} increases as the

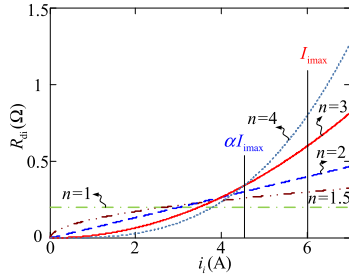


Fig. 4. Droop resistance varying with load current under different droop coefficients, where $n = 1$ denotes a linear droop control as (1).

operating load increases to I_{imax} , while the higher n makes R_{di} much larger in a heavy load range of $[\alpha I_{imax}, I_{imax}]$. In Fig. 4, α depends on the cross point of different curves. However, in the light load range of $[0, \alpha I_{imax}]$, the higher n makes R_{di} much smaller and the linear droop of $n = 1$ has the best current sharing accuracy performance in some light loads. Therefore, a proper n needs to be designed to satisfy the requirement of load current sharing accuracy in a heavy load range of $[\alpha I_{imax}, I_{imax}]$.

By (4), the current sharing index e_I decreases with the incremental R_{di} . Therefore, if (5) is well implemented in a heavy load, the worst case, i.e., the maximum $e_{I_{imax}}$, will not occur at I_{imax} , but an interim load. From Fig. 4, if e_I at the current of αI_{imax} is smaller than $e_{I_{imax}}$ and $n > 1$, e_I will not exceed $e_{I_{imax}}$ during $[\alpha I_{imax}, I_{imax}]$. In this way, a minimum n can be determined in the heavy load.

Taking two parallel-connected converters as an example, according to Kirchhoff's voltage law in Fig. 2, we have

$$v_{bus} = V_N - k_1 i_1^n - R_1 i_1 = V_N - k_2 i_2^n - R_2 i_2. \quad (7)$$

Then it can be further derived as

$$k_1 i_1^n + R_1 i_1 = k_2 i_2^n + R_2 i_2. \quad (8)$$

According to (4), the maximum $e_{I_{imax}}$ should be satisfied. Thus

$$|i_1 - i_2| \leq e_{I_{imax}}(i_1 + i_2). \quad (9)$$

By solving (8) and (9), the expressions in the two cases can be obtained.

1) If $i_1 \leq i_2$, then $i_2 \leq \gamma i_1$.

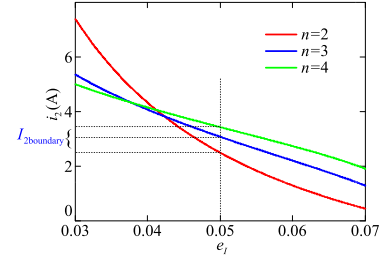
$$\begin{cases} i_1 \leq \frac{n-1}{\sqrt{k_1 - k_2 \gamma^n}} \sqrt{\frac{R_2 \gamma - R_1}{k_1 - k_2 \gamma^n}} & \text{if } k_1 - k_2 \gamma^n > 0 \\ i_1 \geq \frac{n-1}{\sqrt{k_2 \gamma^n - k_1}} \sqrt{\frac{R_1 - R_2 \gamma}{k_2 \gamma^n - k_1}} & \text{if } k_1 - k_2 \gamma^n < 0, \end{cases} \quad (10)$$

where $\gamma = \frac{1+e_{I_{imax}}}{1-e_{I_{imax}}}$.

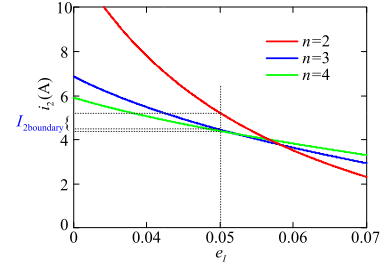
2) If $i_1 > i_2$, then $i_1 \leq \gamma i_2$

$$\begin{cases} i_2 \leq \frac{n-1}{\sqrt{k_2 - k_1 \gamma^n}} \sqrt{\frac{R_1 \gamma - R_2}{k_2 - k_1 \gamma^n}} & \text{if } k_2 - k_1 \gamma^n > 0 \\ i_2 \geq \frac{n-1}{\sqrt{k_1 \gamma^n - k_2}} \sqrt{\frac{R_2 - R_1 \gamma}{k_1 \gamma^n - k_2}} & \text{if } k_2 - k_1 \gamma^n < 0. \end{cases} \quad (11)$$

Selection of case 1 and case 2 is determined by judging the real solutions of i_1 and i_2 with the given parameters.



(a) $R_1 = 0.3 \Omega$ and $R_2 = 0.35 \Omega$



(b) $R_1 = 0.3 \Omega$ and $R_2 = 0.37 \Omega$

Fig. 5. Solved i_2 varying with e_I at the different n for two cases.

To guarantee $e_I \leq e_{I_{imax}}$ during $[\alpha I_{imax}, I_{imax}]$, the solved $I_{1boundary}$ or $I_{2boundary}$ with the n^{th} nonlinear droop function in (10) or (11) must follow

$$\begin{aligned} I_{1boundary} &\leq \alpha I_{1max} \quad \text{for case 1, or} \\ I_{2boundary} &\leq \alpha I_{2max} \quad \text{for case 2.} \end{aligned} \quad (12)$$

With the criteria of (12), the minimum n can be uniquely determined at the given $e_{I_{imax}}$. For example, the converter is designed as Fig. 4. The desired bus voltage regulation is $\pm 2.5\%$, and $e_{I_{imax}}$ is set as 5% . Then $\gamma \approx 1.1$. If $R_1 = 0.3 \Omega$, and $R_2 = 0.32 \Omega$, there is no real solution of $I_{1boundary}$ and $I_{2boundary}$ for both cases, which means a linear droop control is enough with $e_I = 2\%$ by (4). If R_2 is 0.35Ω , by solving (10) and (11), (11) is available with real solutions $I_{2boundary}$ with different n as shown in Fig. 5(a). It can be seen that the smaller n relates to the smaller $I_{2boundary}$, which is consistent with curves in the light load range in Fig. 4. In the range of $[0, I_{2boundary}]$, the linear curve with $n = 1$ has a smaller $e_I = 4.8\%$. If R_2 is 0.37Ω , (11) is available as shown in Fig. 5(b), where a higher n generates a smaller $I_{2boundary}$ to satisfy (12). From Fig. 5(b), if α is set as 0.75 , the minimum $n = 3$ is required.

B. Negative Linear Droop Resistance for Light Load

In the range of $[0, \alpha I_{imax}]$, the linear droop control is usually used to permit the desired voltage regulation. From Fig. 4, the current sharing accuracy of the linear droop control is better than other curves with large droop coefficients in the light load range, although e_I may exceed the required $e_{I_{imax}}$. By (4) and $i_1 + i_2 = i_L$, we have

$$i_{1,2} = \frac{i_L(1 \pm e_I)}{2}. \quad (13)$$

Taking the above example, $R_1 = 0.3 \Omega$ and $R_2 = 0.37 \Omega$. α is set as 0.75 and $I_{1,2max} = 6 \text{ A}$. During $[9 \text{ A}, 12 \text{ A}]$ of the load, a third droop curve is used to permit the voltage

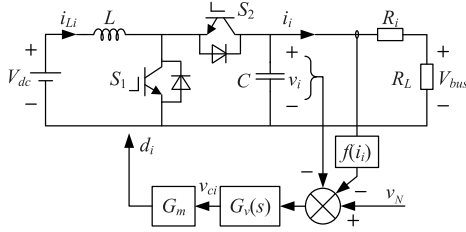


Fig. 6. The schematic of bidirectional buck/boost converter with droop control.

regulation below 2.5% and the current sharing accuracy below 5% from Section II-A. During $i_L \in [0 \text{ A}, 9 \text{ A}]$, the linear droop curve is used to permit the voltage regulation below 2.5%, whilst the current sharing accuracy error e_I can be calculated as 6.5%. By (13), if $i_L = 9 \text{ A}$, the output currents of two converters are about 4.8 A and 4.2 A, which are far below converter ratings. Therefore, to further decrease the voltage regulation at the light load range, we can decrease the linear droop resistance even up to a negative resistance. Obviously, the smaller or even negative droop resistance will lead to a large e_I . By (13), the allowable e_I can be obtained as

$$e_{I_{\text{allow}}} = \frac{2I_{\text{imax}}}{i_L} - 1. \quad (14)$$

So we can define a new linear droop resistance R'_{di} satisfying (14).

$$R'_{\text{di}} = \beta \cdot R_{\text{di}}, \quad (15)$$

where $0 < \beta \leq 1$ denotes a smaller positive droop resistance and $\beta < 0$ denotes a negative droop resistance. It is apparent that a negative droop resistance will have a tighter voltage regulation and lead to a larger e_I .

Also, a negative droop resistance may affect converter stability performance. Taking a bidirectional buck/boost converter as an example, as shown in Fig. 6, the output voltage reference is modified by the droop function $f(i_i) = R'_{\text{di}}i_i$. Here, $G_v(s)$ is the transfer function of the PI controller in the voltage loop, and G_m is the modulation function of the duty cycle. From Fig. 6, the small-signal block diagram of the whole converter is thus demonstrated in Fig. 7, where $G_{vd}(s) = \frac{\Delta v_i(s)}{\Delta d_i(s)}|_{\Delta i_L(s), \Delta v_{dc}(s)=0}$, $G_{vv}(s) = \frac{\Delta v_i(s)}{\Delta v_{dc}(s)}|_{\Delta i_L(s), \Delta d_i(s)=0}$, $G_{id}(s) = \frac{\Delta i_L(s)}{\Delta d_i(s)}|_{\Delta i_L(s), \Delta v_{dc}(s)=0}$, and $G_{ii}(s) = \frac{\Delta i_L(s)}{\Delta i_i(s)}|_{\Delta v_{dc}(s), \Delta d_i(s)=0}$. $Z_{in}(s)$ and $Z_o(s)$ are the input and output impedances of the converter, separately.

To analyze the stability of the converter, the open-loop gain $T(s)$ is required. In Fig. 7, there are two loops with loop gains defined as $T_v(s)$ and $T_i(s)$. By the control theory, the loop gain $T(s)$ can be expressed as

$$T(s) = \frac{T_v(s)}{1+T_i(s)}, \quad (16)$$

where,

$$T_v(s) = G_v(s)G_mG_{vd}(s), \quad \text{and} \quad (17)$$

$$T_i(s) = G_v(s)G_mG_{vd}(s)\beta R_{\text{di}} \frac{1}{Z_o(s)}. \quad (18)$$

Using the PI controller, $G_v(s) = k_{pv} + \frac{k_{iv}}{s}$, and k_{pv} and k_{iv} are proportional and integral coefficients. The modulation function

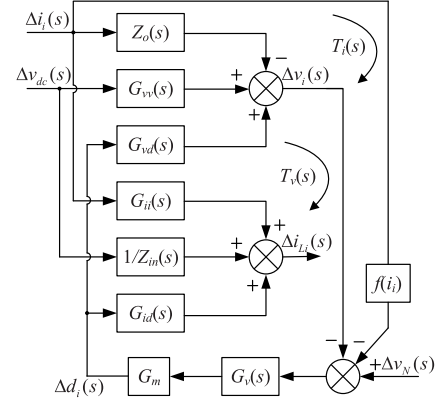


Fig. 7. The small-signal control block diagram of Fig. 6 using the droop control function $f(i_i)$.

G_m is usually expressed as $\frac{1}{V_M}$, where V_M is the amplitude of the ramp signal. To achieve the transfer function $G_{vd}(s)$ and $Z_o(s)$, the state space equations can be established as follows.

$$\begin{cases} L \frac{d\Delta i_{Li}}{dt} = \Delta v_{dc} + V_i \Delta d_i - (1 - D_i) \Delta v_i \\ C \frac{d\Delta v_i}{dt} = (1 - D_i) \Delta i_{Li} - I_{Li} \Delta d_i - \Delta i_i, \end{cases} \quad (19)$$

where D_i represents the steady-state duty cycle of S_1 in the i^{th} converter. From (19), we have

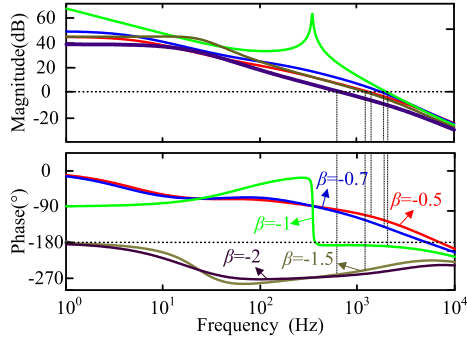
$$Z_o(s) = \frac{Ls}{LCs^2 + \frac{L}{R_L+R_i}s + (1 - D_i)^2}, \quad \text{and} \\ G_{vd}(s) = \frac{V_{dc} - I_{Li}Ls}{LCs^2 + \frac{L}{R_L+R_i}s + (1 - D_i)^2}. \quad (20)$$

Substituting (20) into (17) and (18), $T(s)$ can be plotted in bode curves, as shown in Fig. 8. The converter parameters are listed in Table I in Section IV. From Fig. 8, it shows that a large negative droop resistance corresponds to a large phase margin at the crossover frequency f_c . When β becomes -1.5 or -2 , the converter is even not stable. Therefore, we choose β of -0.7 to have a phase margin of 37.7° . For the example case, the new linear droop control has a tight voltage regulation of 0.44% and the current sharing accuracy of 17.9%, not exceeding $e_{I_{\text{allow}}}$ of 33.3%.

C. Curve Fitting Method for Smooth Transition

According to the above discussion, a segmented curve is utilized to realize the negative linear droop resistance, i.e., $n = 1$, at light loads of $[0, \alpha I_{\text{imax}}]$ and a larger nonlinear positive droop resistance, e.g., $n = 3$, at heavy loads of $[\alpha I_{\text{imax}}, I_{\text{imax}}]$, as the red and blue solid curves shown in Fig. 9. Usually, the least square curve fitting algorithm is required to combine these two curves. Here, to realize the smooth transition, a boost-up of the nonlinear curve reaching the end of the linear curve can be adopted further reducing the voltage drop at heavy loads, as the brown curve in Fig. 9. The brown curve can be expressed as

$$v_{\text{ref}i} = V_N + \sum_{m=1}^n k_m i_i^m, \quad (21)$$

Fig. 8. Magnitude and phase curves of $T(s)$ under different β .

Parameter	Value
Nominal voltage V_N	48 V
Nominal current I_N	6 A
Input voltage V_{dc}	36 V
Line resistance R_1, R_2	0.3Ω, 0.37Ω
Switching frequency f_s	50 kHz
Inductor L	150 μH
Output capacitor C	470 μF

where k_m represents polynomial fitting coefficients, and n indicates the highest order of the fitting curve. By (21), the droop function becomes

$$f(i_i) = \sum_{m=1}^n k_m i_i^m. \quad (22)$$

It can be readily implemented in Fig. 6.

D. Extension to Multiple Parallel-Connected Converters

If the DC microgrid system has multiple parallel-connected converters, the above analysis is still available. In such a case, output currents for N converters can be expressed as

$$\begin{cases} i_1 = \frac{V_N R'_2 R'_3 \cdots R'_N}{R'_1 R'_2 \cdots R'_N + R_L \sum_{j=1}^N \frac{R'_1 R'_2 \cdots R'_N}{R_j}} \\ i_2 = \frac{V_N R'_1 R'_3 \cdots R'_N}{R'_1 R'_2 \cdots R'_N + R_L \sum_{j=1}^N \frac{R'_1 R'_2 \cdots R'_N}{R_j}} \\ \dots \\ i_N = \frac{V_N R'_1 R'_2 \cdots R'_{N-1}}{R'_1 R'_2 \cdots R'_N + R_L \sum_{j=1}^N \frac{R'_1 R'_2 \cdots R'_N}{R_j}} \end{cases} \quad (23)$$

where $R'_j = R_j + R_{dj}$ and $j \in [1, N]$. i_p and i_q are defined as the maximum output current and the minimum output current among N converters, i.e., $i_p = \max\{i_1, i_2, \dots, i_N\}$ and $i_q = \min\{i_1, i_2, \dots, i_N\}$.

Following (4), the maximum current sharing error e_I between i_p and i_q can be given as

$$e_I = \left| \frac{i_p - i_q}{i_p + i_q} \right| = \left| \frac{R_q - R_p + R_{dq} - R_{dp}}{R_q + R_p + R_{dq} + R_{dp}} \right|. \quad (24)$$

If the maximum current sharing error and voltage regulation satisfy the desired requirements, the output currents and voltages of the other converters must be within the desired ranges. Therefore, with (24) and the acceptable bus voltage minimum V_{\min} , the nonlinear droop curve could be designed according to Section II-A, B, and C.

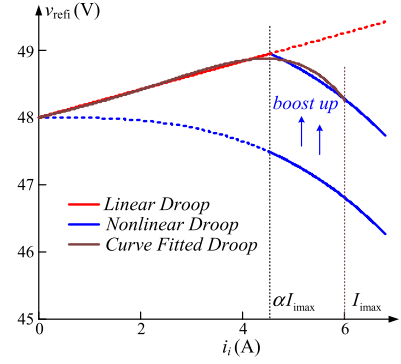


Fig. 9. Segmented droop control curves and their fitted curve.

III. STABILITY ANALYSIS

Although the design of negative droop resistance has considered the system stability issue in Section II-B, the analysis is based on the converter with the constant output power and cable resistances. In practice, the converter may operate at different power levels, and the cable resistances and inductances also vary with temperature and time, as shown in Fig. 10. To check the system stability in such a range of operation conditions, the state variable vector of the system can be got as $\mathbf{x} = [\Delta v_1 \Delta v_2 \Delta i_1 \Delta i_2 \Delta i_{L1} \Delta i_{L2} \Delta v_{c1} \Delta v_{c2}]^T$.

From Fig. 6, it gives

$$v_{ci} = [v_N + f(i_i) - v_i] \left(k_{pv} + \frac{k_{iv}}{s} \right). \quad (25)$$

Upon differentiating both sides of equation in (25), we get

$$\frac{dv_{ci}}{dt} = k_{pv} \left[f(i_i)' \frac{di_i}{dt} - \frac{dv_i}{dt} \right] + k_{iv} [v_N + f(i_i) - v_i], \quad (26)$$

where $f(i_i)' = \sum_{m=1}^n m k_m i_i^{m-1}$.

According to Fig. 10, the state equations of a two-converter based microgrid can be linearized by using (19) and (26). It gives

$$\begin{aligned} \frac{dx_1}{dt} &= -\frac{1}{C}x_3 + \frac{D'_1}{C}x_5 - \frac{I_{L1}G_m}{C}x_7, \\ \frac{dx_2}{dt} &= -\frac{1}{C}x_4 + \frac{D'_2}{C}x_6 - \frac{I_{L2}G_m}{C}x_8, \\ \frac{dx_3}{dt} &= \frac{1}{L_{l1}}x_1 - \frac{R_1 + R_L}{L_{l1}}x_3 - \frac{R_L}{L_{l1}}x_4, \\ \frac{dx_4}{dt} &= \frac{1}{L_{l2}}x_2 - \frac{R_L}{L_{l2}}x_3 - \frac{R_2 + R_L}{L_{l2}}x_4, \\ \frac{dx_5}{dt} &= -\frac{D'_1}{L_1}x_1 + \frac{V_1 G_m}{L_1}x_7, \\ \frac{dx_6}{dt} &= -\frac{D'_2}{L_2}x_2 + \frac{V_2 G_m}{L_2}x_8, \\ \frac{dx_7}{dt} &= -(k_{pv} \frac{dx_1}{dt} + k_{iv}x_1) + f(i_1)'(k_{pv} \frac{dx_3}{dt} + k_{iv}x_3), \\ \frac{dx_8}{dt} &= -(k_{pv} \frac{dx_2}{dt} + k_{iv}x_2) + f(i_2)'(k_{pv} \frac{dx_4}{dt} + k_{iv}x_4). \end{aligned} \quad (27)$$

By (27), the state equations can be expressed as

$$\dot{\mathbf{x}} = \mathbf{A}\mathbf{x}. \quad (28)$$

Then the state matrix \mathbf{A} can be derived as shown in (29), at the bottom of the next page. The eigenvalues of matrix

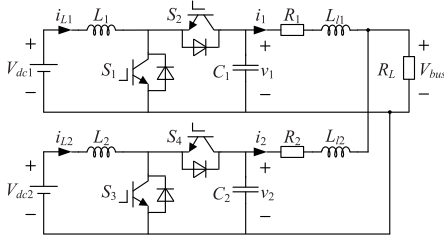


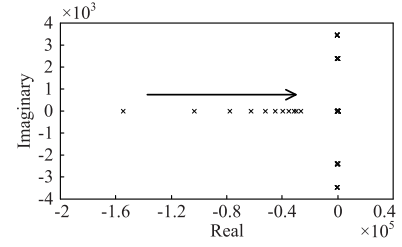
Fig. 10. The simplified DC microgrid model for stability analysis.

A can be calculated by the state-space equation eigenvalue analysis. If all the eigenvalues in matrix A are located in the left half-plane of s domain, the system is stable under the small disturbance near the steady-state operating point [24], [25].

With the above analysis, the root loci of the system eigenvalues can be plotted under different power and cable parameters, as shown in Fig. 11. If the system operates at the fixed cable parameters, i.e., $R_1 = 0.3 \Omega$, $R_2 = 0.37 \Omega$, and $L_{l1} = L_{l2} = 300 \mu\text{H}$, Fig. 11(a) demonstrates the root locus of eigenvalues in a power range from 0 to the full load of 600 W, where all eigenvalues locate in the left half-plane. Thus, the system is stable in the designed power range. With the increasing power, the eigenvalues move to the right half-plane, which indicates the system stability margin becomes small. Fig. 11(b) gives the root locus of the eigenvalues with one increasing cable resistance R_2 from 0.1Ω to 1.5Ω and R_1 keeping as 0.3Ω at a fixed load rating of 600 W as an example. From Fig. 11(b), all eigenvalues locate in the left half-plane and move left with the increasing R_2 , which indicates a stable system in the variation of cable resistance. The root locus of eigenvalues, shown in Fig. 11(c), is obtained by varying the cable inductance from $20 \mu\text{H}$ to $650 \mu\text{H}$, while $R_1 = 0.3 \Omega$, $R_2 = 0.37 \Omega$, and the power is fixed at 600 W. It should be noticed that the real parts of some eigenvalues reach zero when the cable inductance rises to $610 \mu\text{H}$. If the cable inductance continues to increase, some eigenvalues go into the right plane and the system becomes unstable. Therefore, the cable inductances can not be designed so large to avoid the unstable system.

IV. EXPERIMENTAL VERIFICATION

To verify the above analysis, two same bidirectional buck/boost converters with each power level of about 300 W are built and connected in parallel to form a simple DC



(a) with the increasing power.

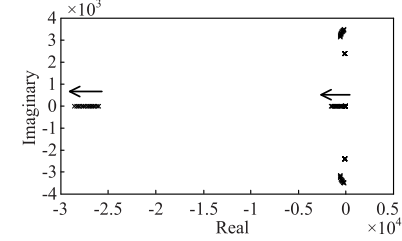
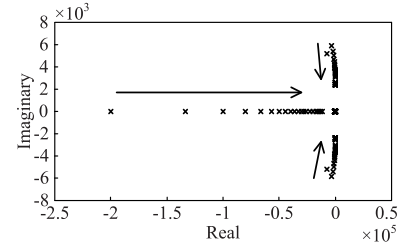

 (b) with the increasing cable resistance R_2 .

 (c) with the increasing cable inductance L_{li} .

Fig. 11. Root loci of system eigenvalues with (a) the increasing power, (b) the increasing cable resistance and (c) the increasing cable inductance.

microgrid. The experimental parameter is shown in Table I. The DC input voltage of two converters is 36 V and the no-load output voltage V_N is set as 48 V. The maximum output current I_{\max} of each converter is 6 A. The desired voltage regulation is $\pm 2.5\%$ and the maximum current sharing accuracy error is 5%. Line resistances are set as $R_1 = 0.3 \Omega$ and $R_2 = 0.37 \Omega$. The fitted droop control function $f(i_i)$ in Fig. 9 is $f(i_i) = 0.1887i_i + 0.00593i_i^3 - 0.000278i_i^5$, where converters 1 and 2 have the same parameters and curves. The control PI parameters of each converter are as follows: $k_{pv} = 0.5$, and $k_{iv} = 200$. α is set as 0.75 and β is set as

$$A = \begin{bmatrix} 0 & 0 & -\frac{1}{C} & 0 & \frac{D'_1}{C} & 0 & -\frac{I_{L1}G_m}{C} & 0 \\ 0 & 0 & 0 & -\frac{1}{C} & 0 & \frac{D'_2}{C} & 0 & -\frac{I_{L2}G_m}{C} \\ \frac{1}{L_{l1}} & 0 & -\frac{R_1+R_L}{L_{l1}} & -\frac{R_L}{L_{l1}} & 0 & 0 & 0 & 0 \\ 0 & \frac{1}{L_{l2}} & -\frac{R_L}{L_{l2}} & -\frac{R_2+R_L}{L_{l2}} & 0 & 0 & 0 & 0 \\ -\frac{D'_1}{L_1} & 0 & 0 & 0 & 0 & 0 & \frac{V_1G_m}{L_1} & 0 \\ 0 & -\frac{D'_2}{L_2} & 0 & 0 & 0 & 0 & 0 & \frac{V_2G_m}{L_2} \\ \frac{k_{pv}f(i_1)'}{L_{l1}} - k_{iv} & 0 & a_{73} & -\frac{k_{pv}f(i_1)'}{L_{l1}} & -\frac{k_{pv}D'_1}{C} & 0 & \frac{k_{pv}I_{L1}G_m}{C} & 0 \\ 0 & \frac{k_{pv}f(i_2)'}{L_{l2}} - k_{iv} & -\frac{k_{pv}f(i_2)'}{L_{l2}} & a_{84} & 0 & -\frac{k_{pv}D'_2}{C} & 0 & \frac{k_{pv}I_{L2}G_m}{C} \end{bmatrix},$$

where $a_{73} = \frac{k_{pv}}{C} + k_{iv}f(i_1)' - \frac{k_{pv}f(i_1)'(R_1 + R_L)}{L_{l1}}$ and $a_{84} = \frac{k_{pv}}{C} + k_{iv}f(i_2)' - \frac{k_{pv}f(i_2)'(R_2 + R_L)}{L_{l2}}$ (29)

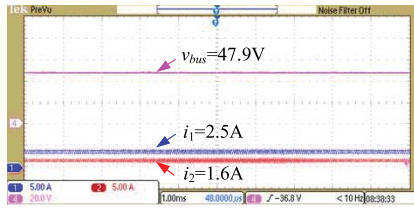


Fig. 12. Waveforms of v_{bus} , i_1 and i_2 at a light total load of 200 W.

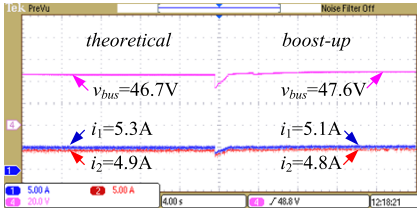


Fig. 13. Waveforms of v_{bus} , i_1 and i_2 at a heavy load of 480 W with the theoretical 3rd droop coefficient (left) and the boost-up fitted curve in Fig. 9 (right).

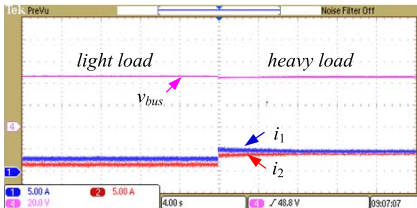


Fig. 14. Dynamic waveforms of v_{bus} , i_1 and i_2 at a load stepping from 200 W to 480 W.

–0.7. The control scheme is carried out in a TMS320F28335 microcontroller.

Fig. 12 shows the experimental waveforms of the bus voltage v_{bus} , converter output currents i_1 and i_2 at a light total load of about 200 W. Here, with the negative droop resistance, v_{bus} is measured as 47.9 V and the voltage deviation e_V is calculated as 0.2% far below the limit of $\pm 2.5\%$. The current sharing accuracy e_I can be calculated as 21.9%, which is consistent with the theoretical 17.9%. The difference is caused by the fitted curve, which is not the exactly linear line. Although the experimental e_I is far beyond the required 5%, it is designed below $e_{I,allow}$ of 33.3% by (14). Thus two converters work in the safe current range and show a tight voltage regulation.

When the load is increased to 480 W beyond α of total power 600 W, a higher droop coefficient n of 3 is used. The left side in Fig. 13 shows the experimental waveforms of v_{bus} , i_1 and i_2 with the theoretical 3rd droop coefficient curve. From the measured v_{bus} , i_1 and i_2 , e_V is calculated as 2.7%, close to the predesigned 2.5%. The small tolerance is usually caused by measurement error. And e_I is measured as 3.9%, satisfying the required 5%. So the 3rd droop coefficient curve is proved to realize the desired voltage regulation and current sharing accuracy in a specified heavy load range. To further tighten the voltage regulation, a boost-up 3rd droop coefficient curve, i.e., the boost-up curve in Fig. 9, is implemented and shows the waveforms on the right side of Fig. 13. Due to the fitted boost-up curve, e_I becomes 3%, a bit different from the initial 3.9%. But the bus regulation can be well improved at 0.8%. So the fitted curve shows good performance in both voltage

regulation and current sharing. Fig. 14 also gives the dynamic waveforms with load changing from 200 W to 480 W. The proposed nonlinear droop control scheme ensures a fast and stable system.

V. CONCLUSION

To permit a DC microgrid with the desired bus voltage regulation and current sharing accuracy, this article proposes a method to identify the nonlinear droop curve coefficient in a specified heavy load range, while a linear droop control with negative droop resistances is used to tighten the bus voltage regulation under the allowable current sharing accuracy in the light load range. The design procedure of load boundary and the effect of negative droop resistances on the system stability are detailed. Besides, a boost-up and curve-fitting method can further improve the voltage regulation and achieve the smooth transition between light load and heavy load. In this way, no additional voltage compensation loop is needed. The experimental results verify the theoretical analysis well. The method can be readily extended to the DC microgrid with multiple parallel-connected converters.

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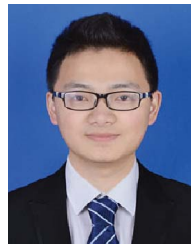
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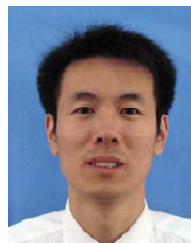
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